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Northern  
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Report of  
Investigation, Red  
Lodge/Bearcreek  
subsidence



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Engineering  
and Testing, Inc.

REPORT OF INVESTIGATION  
RED LODGE/BEARCREEK  
SUBSIDENCE POTENTIAL STUDY

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**Northern**

Engineering  
and Testing, Inc.

600 South 25th Street  
P. O. Box 30615  
Billings, MT 59107  
(406) 248-9161

October 30, 1987

State of Montana  
Department of State Lands  
Abandoned Mine Reclamation Bureau  
1625 11th Avenue  
Helena, Montana 59620

Attention: Mr. Dick Juntunen

Subject: Report of Investigation Red Lodge/Bearcreek  
Subsidence Potential Study  
Project No. 87-3001.D

Gentlemen:

In accordance with our Professional Services Agreement dated January 1, 1987 and Work Plan dated March 20, 1987 with revisions dated May 19, 1987 and August 31, 1987 we have performed a subsidence potential study for abandoned coal mines in the Red Lodge/Bearcreek area of Montana. The purpose of this study was to identify locations where the potential for subsidence can be considered low, medium or high. We understand this information will be utilized by the Abandoned Mine Reclamation Bureau for determining if a subsidence insurance program is desirable.

The study primarily focused on the potential for subsidence at presently developed locations such as business and residential areas and public roads. The potential for subsidence at adjoining areas which are undeveloped or used for agricultural purposes was considered using empirical relationships rather than analytical prediction techniques.

One relatively shallow (less than 100 feet deep) adit was encountered extending under Highway 308 between Washoe and Bearcreek near the present entrance to the Red Lodge Coal Company. In our Work Plan Revision dated August 31, 1987, we proposed to backfill this adit. A separate report will be submitted detailing our recommendations for backfilling. No other locations of relatively shallow workings were identified beneath presently developed areas.

The analysis we performed considered the following modes of subsidence:

- 1) Chimney Subsidence
- 2) Roof Collapse (Beam Theory)
- 3) Pillar Crushing
- 4) Pillar Punching



State of Montana  
Attn: Mr. Dick Juntunen  
Helena, Montana

October 30, 1987  
Page Two

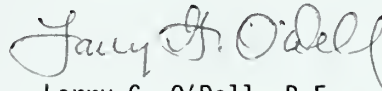
No single failure mode is favored; the most probable mode changes from one seam or location to another. Our analysis indicates at most locations the critical load condition for modes 2, 3 and 4 occurred while the mines were in operation, because mine dewatering caused roofs and pillars to support saturated overburden while under present conditions most areas support the buoyant weight of overburden. The calculated factor of safety for those failure modes is estimated to have increased by a factor of about 1.7 following closure and aquifer recharge into the mines.

The subsidence potential beneath presently developed areas is generally low. Exceptions to this generalization are identified in the report.

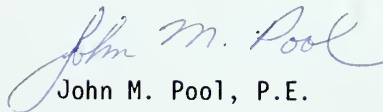
The following report presents the investigations and analysis in detail, and maps in the Appendix show our interpretation of subsidence potentials.

We sincerely appreciate the opportunity to conduct this interesting and challenging study. Should you have any questions regarding this report please contact us at your convenience.

Respectfully submitted,



Larry G. O'Dell, P.E.



John M. Pool, P.E.

LGO/JMP/rmr





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## APPENDIX A - Plate Nos. 1 through 6

Drill Logs followed by Photographs

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Cross Sections of Mine Limits	(87-3001.D-7)

## APPENDIX B - Calculations: Beam Analysis and Pillar Crushing



## I. INTRODUCTION

The objectives of our work scope were :

1. Develop a subsidence potential map for the town of Red Lodge
2. Prepare plans and specifications for remedial alternatives where shallow workings extend under Highway 308.

During the course of our study these two objectives were modified slightly to limit remedial alternatives to locations where workings under Highway 308 are shallow, and extended the subsidence potential study area to include the Bearcreek area.

The Red Lodge/Bearcreek Project is located in south central Montana. Underground mining began in the 1880's and substantially ceased in the 1930's. A few mines continued operating until the 1950's when virtually all mining ceased.

Within the Red Lodge area a total of seven seams were mined, resulting in multiple extractions beneath much of the area. Detailed mine maps were kept and are now the property of Meridian Minerals Company.

Within the Bearcreek area documentation of mine locations is poorly defined. Although some maps exist, ties to surface reference points were generally not made. Observation of the available maps indicates multiple seam extractions were not made.

A more thorough presentation of the mining history was presented by others (4) in a previous study. Since that subject was not part of our work scope it is not repeated in this report.

## II. GENERAL GEOLOGY

The geologic profile in the Red Lodge area typically consists of coarse grained alluvium overlying the Fort Union Formation. The thickness of alluvium is variable, ranging from a few feet to about 100 feet. It consists of gravel, sand, cobbles and boulders. A somewhat similar profile exists in the Bearcreek area except the materials overlying the Fort Union Formation are a combination of colluvium and alluvium and are predominately fine grained sand and clay.

The Fort Union Formation consists of sequences of sandstone, claystone and coal beds. It is locally tilted such that it dips toward the south at angles between 16 to 19 degrees near Red Lodge and at less than 5 degrees in the Bearcreek area.

The average thicknesses and distances between the various coal seams were obtained from cross sections belonging to Meridian Minerals, and are as follows:



<u>Seam No.</u>	<u>Average Thickness, ft.</u>	<u>Depth to Next Seam, ft.</u>
1	9	105
1 1/2	5.5	65
2	8	70
3	10	118
4	10	175
5	8	30
6	6	N/A
Bearcreek	8.5	N/A

Specific conditions at each boring location are described later in this report, and individual boring logs are presented in Appendix A.

### III. INVESTIGATIONS

The investigations consisted of the following activities:

- Study of existing mine maps and observation of surficial conditions.
- Subsurface explorations utilizing rotary and coring techniques.
- Borehole photography where practical.

A proposed resistivity survey was not performed after we determined that workings along Highway 308 would likely be deeper than could be reliably detected by that method.

#### A. Map and Surficial Conditions Study

Maps of the mines were borrowed from Meridian Minerals. The maps show that room and pillar extraction methods were used. It also appears that pillars were robbed from some areas, probably during retreat. Those maps were used to prepare Drawing Nos. 87-3001.D-1 through 4, in Appendix A, which show the depth of cover and cumulative thickness of coal mined from beneath specific locations. These were then used to create the final subsidence potential maps Drawing Nos. 87-3001.D-5 & 6, showing the ratio of depth of cover to mined thickness. Empirical correlations have been developed (7) which utilize that ratio to approximate the subsidence potential. A preliminary subsidence potential map was prepared to help target general locations for detailed subsurface explorations. Maps of individual seams were then studied to select specific hole locations. Our intent was to intersect rooms or adits, and where multiple seams were mined, to intersect rooms in more than one seam. From mine floor elevations shown on the maps, three cross sections were constructed showing the orientations of mined areas with respect to one another and to the ground surface. These are presented on Drawing No. 87-3001.D-7.

#### B. Subsurface Explorations

The following seven locations were ultimately selected for subsurface explorations:



- Red Lodge Airport
- Highway 308 just east of Red Lodge
- Red Lodge - 14th Street between Hauser and Main
- Washoe Adit under Highway 308
- Washoe Mine adjacent to Highway 308
- Smith Mine No. 2 Adit under Highway 308
- Smith Mine No. 2 Unmapped Adit under Highway 308

Locations of these borings are shown on the drawings in Appendix A. Drilling was performed by Rock Creek Drilling under the supervision of our personnel. The following paragraphs summarize the conditions encountered at each location. Logs of the borings are presented in Appendix A.

After drilling was complete a video tape was made of the boreholes, where practical. The photography met with varying degrees of success. Two factors limited camera use; 1) water turbidity and 2) an underwater depth limitation of 100 feet.

### Red Lodge Area

At the airport, DH-1 intersected both the No. 1 1/2 and No. 2 Seams. While approaching the No. 1 1/2 Seam evidence of roof strain and possible collapse began to appear below 117.5 feet and continued down to the coal at 129 feet. The No. 1 1/2 Seam is between 129 and 134 feet; it is highly fractured. We interpret the zone between 117.5 to 129 to be fractures opened by roof collapse. The entire thickness of the coal was present, as shown by the video tape. This indicates a pillar rather than a room was intersected.

The hole continued down to the No. 2 Seam near 204 feet. The rate of drill penetration increased below 200 feet but the video tape does not show a change in the rock. A definite void is present between 204 and 209 feet; then about seven feet of soft broken material is present before the mine floor is encountered. The camera could not progress deeper than 209 feet due to the broken rock. The rubble is interpreted to be roof fall material. The No. 2 Seam averages eight feet thick so it appears roof collapse has progressed only about four feet above the original roof. The increased rate of drill penetration below 200 feet might indicate a weakening of the rock, but it could not be detected by the camera.

On the east bench DH-2 was drilled in an area where mine maps show the No. 1 1/2 Seam was extensively robbed of pillars. This location was selected for the purpose of determining how far upward roof collapse might extend, for comparison with empirical correlations. Our depth of cover map predicted the floor to be close to 350 feet. Continuous coring began at 310 feet and continued to 360 feet. Evidence of roof strain and collapse was noted in the core beginning about 326 feet, where open joints began to appear. Some loss of circulation was encountered but no distinct voids detected. Recovery was good and RQD was high throughout the cored interval. Only a two or three foot interval near 345 feet produced highly broken rock. This suggests total convergence of roof and floor has occurred at this location. A





thin layer of coal and partial circulation loss near 358 feet is interpreted to be the mine floor. The hole was continued down to 370 feet with a tricone rotary bit with no additional evidence of collapse or open joints noted below 360 feet. Turbidity and the camera's depth limitation limited photography to a depth of only 175 feet. Within that depth no evidence of subsidence was noted although several fractured and eroded zones are evident.

On 14th Street, DH-3 encountered a pillar in the No. 4 Seam between about 228 and 237 feet. The core from that interval consists of highly fractured coal. The fracturing was so extensive that laboratory strength tests were not possible. Core recovery above the coal was good; about 95 percent with an RQD of 85 percent, indicating the roof to be relatively intact.

The boring was continued down to intersect the No. 5 Seam near 435 feet. A distinct void was not encountered, but roof strain was evident in the core below 425 feet. Absence of coal in the core verifies that a former room was intersected. Since neither a distinct void nor a significant thickness of rubble was detected, we believe there has been roof-floor convergence, probably due to pillar crushing or punching. The roof strain noted in the core suggests pillar crushing or punching has caused the roof to sag. Borehole photography was not performed due to the camera's underwater depth limitation.

#### Washoe and Smith Mines

A total of four borings were originally planned at these two sites. Two borings (DH-5 and 6) were made at the Washoe Mine near the locations proposed in our Field Investigation Plan dated May 19, 1987. The holes extended to depths of about 250 and 265 feet. DH-5 attempted to intersect the sloped entry but missed and DH-6 intersected a pillar. In both borings about eight feet of coal was encountered very close to the depths predicted by our depth of cover map. Based upon projections of mine map elevations the Washoe sloped entry should be about 200 feet below Highway 308. Neither boring encountered any evidence to suggest subsidence.

At the Smith Mine, two borings (DH-7 and 8) were planned to intersect adits under Highway 308. West of the present Red Lodge Coal Company entrance two sags are evident in the pavement surface and correspond to mapped adit locations. At this location three holes (DH-7, 7A and 7B) were drilled attempting to intersect an adit. All three borings encountered a similar profile which we interpret to be fill to about 16 feet, then coal and bedrock. It appears the adits were exposed and backfilled during highway construction. The sags most likely represent differential settlement between the fill and natural materials.

East of the Red Lodge Coal Company entrance an adit not shown on the mine maps exists; it is shown on the Montana Highway Department construction drawings where it is identified as a mine shaft. Two drill holes (DH-8 and 9) were required to intersect this adit. The roof is at a depth of 57.8 feet and a void about 9 feet high is present. About seven feet of well cemented sandstone forms the roof. Photography of this hole showed no evidence of roof strain nor was rubble observed on the floor. Turbidity limited sight distance to about one foot so no attempt was made to view in a horizontal direction.



## C. Laboratory Testing

Approximately 93 percent of the 154 feet of rock cored was recovered. The following laboratory testing program was preformed to obtain representative material properties for analytical analysis.

Drill Hole No.	Sample Depth, ft.	Unconfined Compression	Splitting Tensile Strength	Unit Weight and Moisture Content
1	123.0 - 124.0		X	X
1	135.0 - 136.0	X		X
1	136.0 - 137.0	X		X
2	311.2 - 312.0			X
2	317.0 - 317.8			X
2	333.4 - 334.1			X
2	344.8 - 345.5			X
2	352.5 - 353.3			X
3	212.0 - 213.0	X		X
3	213.0 - 214.0		X	X
3	225.4 - 226.3	X		X
3	227.6 - 228.3		X	X
3	414.0 - 414.7	X		X
3	414.7 - 415.1		X	X
3	441.2 - 442.0	X		X
N/A	Coal	X		X
N/A	Coal	X		X
Totals:		8	4	17

None of the coal recovered from the drill holes produced samples of sufficient size for laboratory strength tests. Several pieces of coal were provided by the Red Lodge Coal Company, which we understand originated from the Smith No. 2 Seam. From these we were successful in cutting two cubes for compressive strength tests.

Although ASTM D2938 does not require recording axial deformation for unconfined compression of rock cores, we recorded that information so that an approximation of the elastic modulus could be calculated. Plots of these tests are shown on Plates 1 through 6 in Appendix A. The table on page 6 summarizes all of the test results.

## IV. ANALYSIS

Having reviewed relevant literature and in consideration of the subsurface conditions encountered, we selected the following methods for analyzing the various failure modes:





SUMMARY OF LABORATORY TEST RESULTS  
RED LODGE/BEARCREEK SUBSIDENCE STUDY  
RED LODGE, MONTANA

Drill Hole No.	Depth in Feet	Classification	Unconfined Compressive Strength, ( $Q_u$ ), ksf	Splitting Tensile Strength, ( $Q_t$ ), ksf	Dry Unit Weight, pcf	Moisture Content, %	Modules of Elasticity (E), psi
DH-1	123.0-124.0	Claystone		50	142	5	
	135.0-136.0	Sandstone	470		146	4	3.5 x 10 <sup>5</sup>
	136.0-137.0	Sandstone	350		147	4	4.2 x 10 <sup>5</sup>
DH-2	311.2-312.0	Sandstone			145	3	
	317.0-317.8	Sandstone			150	2	
	333.4-334.1	Sandstone			147	2	
	344.8-345.5	Claystone			147	3	
	352.5-353.3	Sandstone			142	2	
DH-3	212.0-213.0	Claystone	1020		152	2	9.6 x 10 <sup>5</sup>
	213.0-214.0	Claystone		98	153	3	
	225.4-226.3	Claystone	720		151	2	6.9 x 10 <sup>5</sup>
	227.6-228.3	Claystone		82	151	2	
	414.0-414.7	Sandstone	410		138	5	4.5 x 10 <sup>5</sup>
	414.7-415.1	Sandstone		69	145	2	
N/A N/A	441.2-442.0	Sandstone	1000		151	2	1.0 x 10 <sup>6</sup>
	N/A	Smith No. 2 Coal	401		74	11	
	N/A	Smith No. 2 Coal	310		74	10	



<u>Failure Mode</u>	<u>Method and Source</u>
Chimney Subsidence	Empirical: Karfakis; Subsidence over Abandoned Coal Mines
Roof Collapse	Analytical: Wright; Roof Control Through Beam Action and Arching
Pillar Crushing	Analytical: Bieniawski; Rock Mechanics Design in Mining and Tunneling
Pillar Punching	Analytical: Rockaway and Stephenson; Support of Coal Pillars

Based on a series of assumed mine room failures, the predicted geometry of "trough" shaped subsidence features was estimated using methods presented by Peng; Coal Mine Ground Control. That calculation was developed for use in longwall rather than room and pillar extractions, but it is considered a reasonable approximation for complete roof-floor convergence from any failure mode which does not create significant bulking.

The geometry of this coal field includes two aspects not normally considered in most analytical models: 1) multiple seam extractions and 2) inclined beds. The following discussions of the various failure modes include a description of how we considered those two geometric aspects.

In the interest of brevity, the results of our observations and analysis are summarized first, followed by detailed discussions. The summary statements apply equally for the Red Lodge and Bearcreek areas.

<u>Failure Mode</u>	<u>Summary of Analysis</u>
1. Chimney Subsidence	<ul style="list-style-type: none"> <li>a. Appears to be the mode responsible for previous subsidence features.</li> <li>b. Only the Bill Palmer property east of Red Lodge, a small area west of the end of 10th Street, and two small areas in the Bearcreek area are considered to have a high potential for future chimney subsidence.</li> </ul>
2. Roof Collapse	<ul style="list-style-type: none"> <li>a. Evidence of roof strain observed in the drill holes is limited to within 20 feet or less of the predicted mine roof elevations.</li> <li>b. Calculations indicate the potential for massive roof collapse to be low.</li> <li>c. Using the beam analysis, the roof of Seam No. 5 is calculated to fail about 15 feet above the mine roof, where a competent layer capable of bridging the span was encountered. The other seams were predicted to have very little roof failure.</li> </ul>





- d. For most of the seams, the critical load condition for roof collapse, pillar crushing and pillar punching was during mining because subsequent mine flooding has reduced overburden pressures from saturated to buoyant conditions.

- 3. Pillar Crushing
  - a. Calculations indicate that beneath presently developed areas future pillar crushing is not likely.
  - b. Mines extended deeper than calculations predict pillars should have supported; indicating the coal might be stronger than tests indicate.
- 4. Pillar Punching
  - a. Pillars are not predicted to punch more than one or two feet, due to underlying competent material.
  - b. A short punch distance could initiate the limited roof collapse observed in the borings.

#### A. Chimney Subsidence

Chimney subsidence is one form of roof collapse. It is a progressive upward movement of a void. This failure mode is favored where rock units are thinly bedded. The thin beds constitute thin beams spanning across a void. The collapse of one reduces support for the one above it.

Figure 1 on page 9 shows the three outcomes of chimney subsidence; it can be arrested by a competent layer, by bulking or form a sinkhole at the surface. For this analysis it is assumed that caving will not be arrested by a competent layer; that is considered in the roof collapse discussion. As illustrated in Figure 1 the collapsed material contains voids and therefore occupies a greater volume than the intact material. Given a sufficient overburden thickness this bulking can effectively fill the void. Figure 2 presents heights of caving ratios for various void geometries and bulking factors. A bulking factor in the range of 0.2 has been inferred as an average value for western coal fields from observations by others(2 & 3). Using that value the maximum height of collapse for this mode is in the range of 5 to 15 times the mined seam thickness, for horizontal seams. For the case of a dipping seam it is possible for conical and elliptical chimneys to progress farther. Those two geometries involve limited areas rather than entire rooms. As a result a sloping floor allows material to initially roll away from the collapse location, so a larger area of the room fills with collapsed material before bulking can begin to fill the immediate chimney location. The additional volume of material cannot be easily quantified because it is a function of several variables including horizontal dimensions of the chimney, floor inclination, and location within the room. For the worst case conditions our calculation indicates the volume of additional material flowing out into an inclined room can be about twice that for a horizontal room. Consequently, for worst case conditions, it is possible for conical and elliptical chimneys having small horizontal dimensions to progress nearly twice as far as indicated by Figure 2. However, small chimneys are more prone to being arrested by a more competent layer and therefore rarely reach the surface.



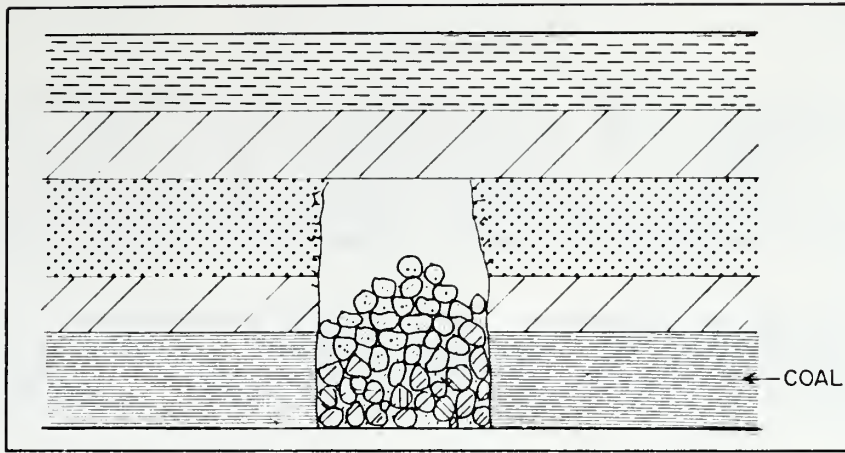


Figure 1a Caving Arrested by a Competent Stratum

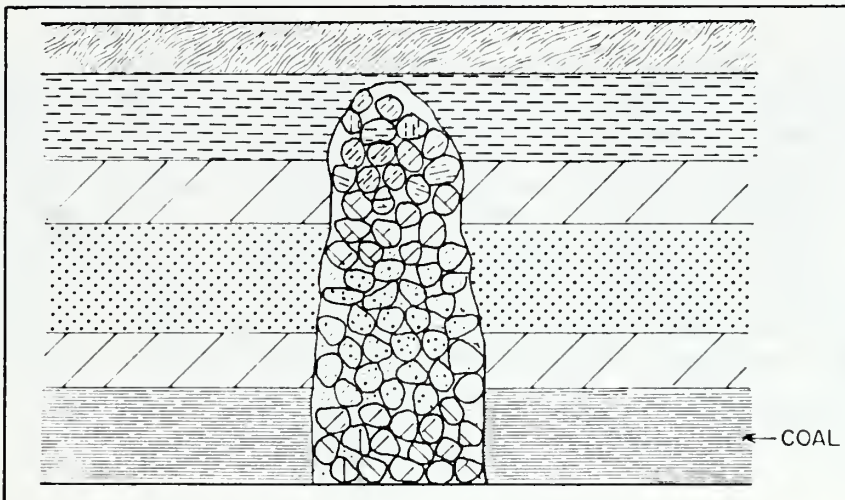


Figure 1b Arrested by Bulking of Roof Debris

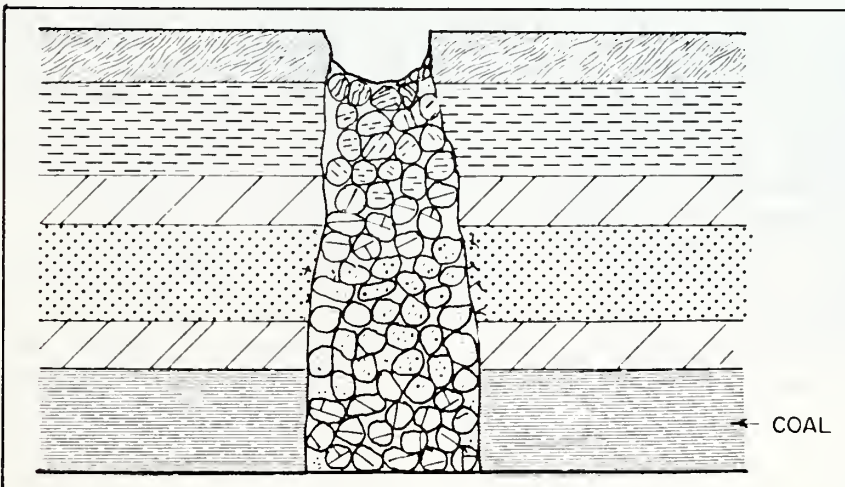


Figure 1c. Formation of a Sinkhole at the Surface

Figure 1 Chimney Subsidence Development



DH-2 was intentionally located where maps show extensive pillar robbing. For that area, we would expect the prism geometry of chimney subsidence to most accurately represent the height for collapse to progress. From the condition of the core and projected mine roof elevation, it appears the height of collapse was limited to 20 feet or less. Based on an average seam thickness of 5.5 feet, Figure 2. suggests a bulking factor in the range of 0.3 for that area.

In consideration of the conditions encountered in DH-2 and other borings, along with the special circumstances that could allow conical and elliptical chimneys to propagate farther than indicated by Figure 2, we selected the following criteria to assign subsidence potential:

<u>Overburden to Mined Thickness Ratio</u>	<u>Subsidence Potential</u>
10:1 or less	High
10:1 to 20:1	Moderate
Greater than 20:1	Low

To consider the influence of multiple seams, one begins with an assumed chimney in the deepest seam. If the distance from that seam up to the next is greater than the predicted collapse height, then that lowest seam can be removed from further consideration. If the contrary is true the upward progression can continue. Iterations continue using the summation of mined thicknesses and separations between seams. Using the cross sections shown on Drawing No. 87-3001.D-7 the following conclusion can be made regarding multiple seam chimney subsidence. The only locations where the potential for future chimney subsidence is considered high are on the Bill Palmer property east of Red Lodge, a small area west of the end of 10th Street in Red Lodge, and where adits are shallow in the Bearcreek area. Those areas are shown on the subsidence potential map as having a high potential for future subsidence. Two small areas within the town and part of the runway classify as having a moderate subsidence potential. A reduction of the Red Lodge Subsidence Potential map is presented on page 12.

## B. Roof Collapse

The mechanism of mine roof collapse is simply a large scale version of chimney subsidence. The concept of the theory is illustrated in Figure 3. The rock layers above the mine are analyzed as a series of beams. Beam thickness is dependent upon thickness of major rock types or bedding or discontinuity characteristics. Each beam is assigned a unit weight, tensile strength and elastic modulus. Initially calculations are made to estimate the deflection of each beam for determining if beams act independently or as composites. In a system of several beams both independent and composite beams can exist in the profile. A sample calculation is presented in Appendix B along with results of specific cases analyzed. Profiles at specific hole locations were modeled using three or four beams. The first beam consisted of the alluvium, having weight but contributing no strength. Subsequent beam dimensions and material properties were selected based upon the log of each hole and laboratory test results. The idealized profiles and material properties analyzed are presented in Figure 4.



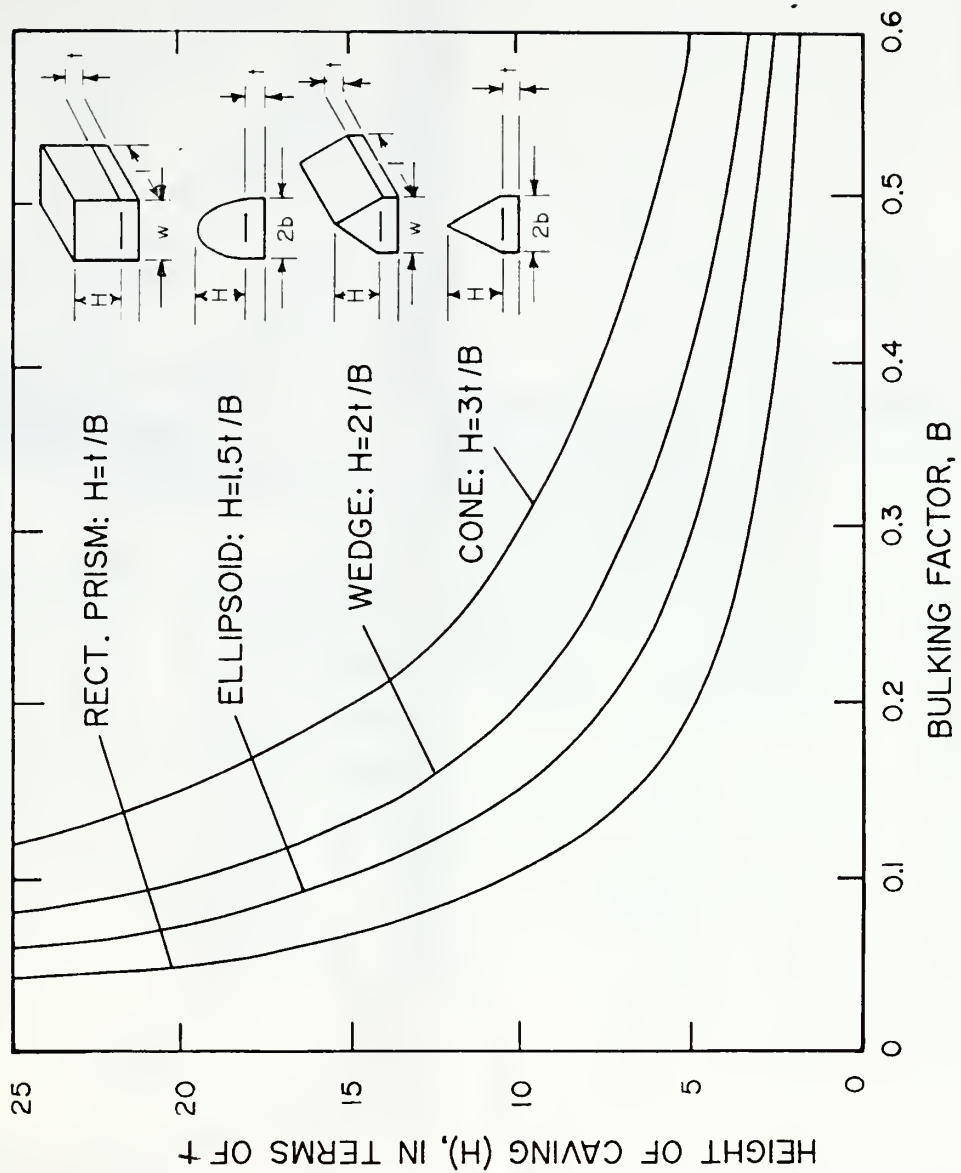


Figure 2 Maximum Height of Stopping

(from Karfakis)







DEPTH OF COVER TO MINED THICKNESS RATIOS

10:1 or less	HIGH SUBSIDENCE POTENTIAL
10:1 - 20:1	MODERATE SUBSIDENCE POTENTIAL
Greater than 20:1	LOW SUBSIDENCE POTENTIAL
Diagonal lines	INSUFFICIENT DATA

STATE OF MONTANA  
DEPARTMENT OF STATE LANDS  
HELENA, MONTANA  
RED LODGE / BEARCREEK  
SUBSIDENCE STUDY  
SUBSIDENCE POTENTIAL MAP

Northern  
Drawing: LNR  
Checked: JMP  
Date: 10/29/87  
Drawing No. 87-300LD-3

RED LODGE SUBSIDENCE POTENTIAL MAP

SCALE 0 500 1000 FEET  
1:50 1:1000

R 20 E



T  
7  
S

35

36

31

32

T  
7  
S

T  
8  
S

2

6

T  
8  
S

T  
9  
S

11

12

7

T  
9  
S

R 19 E

R 20 E

BEARCREEK SUBSIDENCE POTENTIAL MAP

SCALE 0 500 1000 2000 FEET

R 21 E -12A-

APPROXIMATE LIMIT OF MINING

FAULT LINE

APPROXIMATE LIMIT OF MINING

FAULT LINE

DH-6

DH-5

ENTRY TUNNEL

DH-7, 7A, 7B

DH-8

DH-9

BEARCREEK HIGHWAY

Typical Thickness  
in Bearcreek Area  
8-12 feet

DEPTH OF COVER TO MINED THICKNESS RATIOS	
10:1 or less	HIGH SUBSIDENCE POTENTIAL
10:1 - 20:1	MODERATE SUBSIDENCE POTENTIAL
Greater than 20:1	LOW SUBSIDENCE POTENTIAL
	INSUFFICIENT DATA

STATE OF MONTANA  
DEPARTMENT OF STATE LANDS  
BEARCREEK/RED LODGE  
SUBSIDENCE STUDY  
SUBSIDENCE POTENTIAL MAP



Northern  
Engineering and Surveying, Inc.

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Checked: JMP  
Date: 10/29/87

DRAWING NO.  
87-30310-6





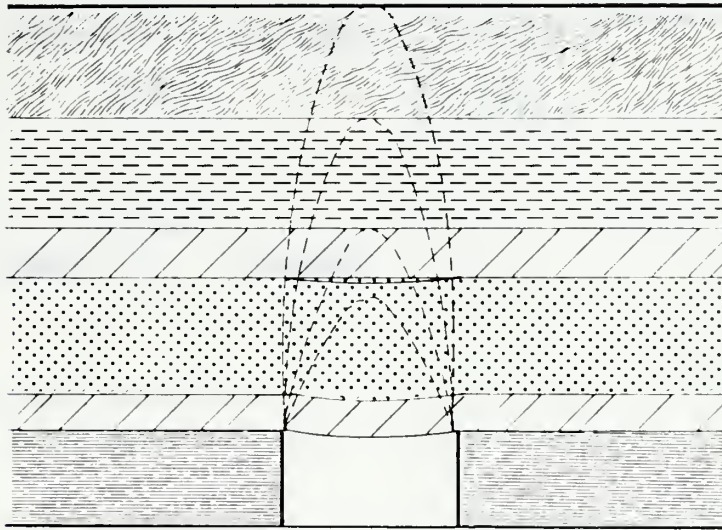
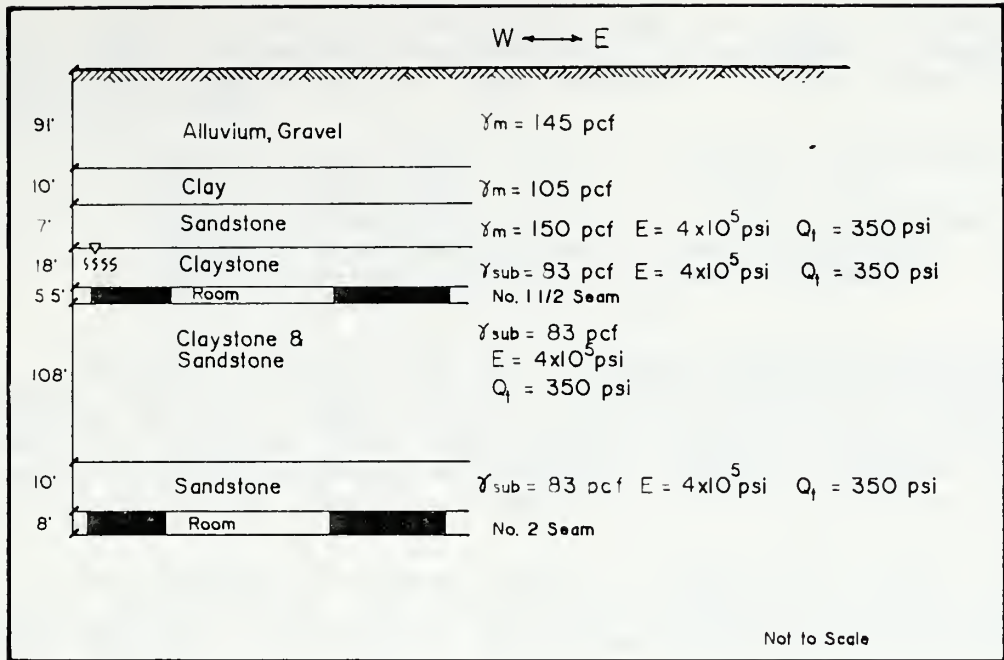
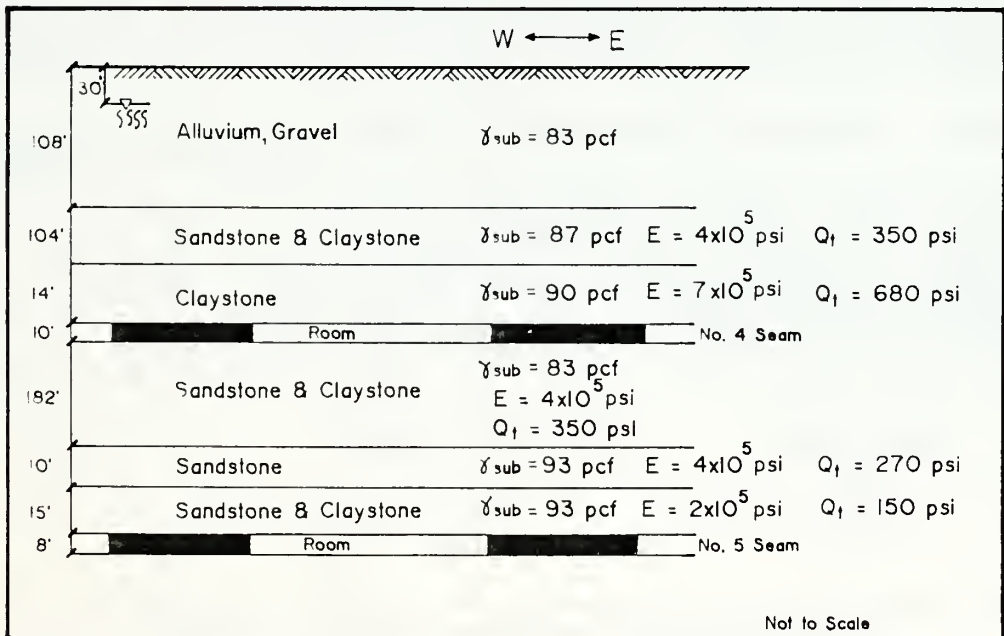


Figure 3 Deflection of the Roof Beams, Bed Separation, and Roof Stress Variation.





4a. DH-1, No. 1-1/2 & No. 2 Seams



4b. DH-3, No. 4 & No. 5 Seams

Figure 4 - Profiles for Roof Collapse Analysis





After an initial calculation was made the input variables were modified to check the analysis for sensitivity to variations in material properties, mine room widths and beam thickness. From this exercise we determined the following:

- The calculated safety factor is inversely proportional to changes in unit weights.
- The calculated safety factor is directly proportional to changes in beam thickness, tensile strength and modulus.
- Variations in room width have an inverse exponential ( $1/x^2$ ) effect on the calculated safety factor.

The unit weight of materials will vary dramatically depending upon groundwater levels and assumptions made regarding hydrostatic conditions. During mining the rooms were dewatered so the saturated unit weight of overburden was supported. After the mines were closed, those portions which are flooded now support the buoyant unit weight of submerged overburden. This change in unit weights generally increases the calculated safety factor by a factor of about 1.7.

The condition of dipping seams was determined to have little influence on the calculated safety factors. To analyze the condition of multiple extractions, only the rock units between the openings were considered in the beam strength.

In order to analyze a variety of conditions, we developed a spreadsheet computer program for roof beam analysis. The computer program used does not distinguish between individual and composite beam theories, it calculates both. In reviewing the computer generated results in Appendix B, the lowest of the two safety factors listed will represent the applicable theory.

In this analysis failure of all beams must occur before subsidence is expressed on the surface.

A determination of the minimum beam thickness required to span across rooms was made using a calculated safety factor of 1.5 as the limiting value. The value of 1.5 was selected as it allows a reasonable variation in material properties, and we believe those assigned to be slightly conservative. The following table presents the minimum lower beam thickness to provide a calculated safety factor greater than 1.5 for each of the cases illustrated in Figure 4.

<u>Seam No.</u>	<u>Beam Thickness, ft.</u>	<u>Safety Factor</u>
1 1/2	1	1.7
2	1	1.9
4	0.5	1.7
5	1.5*	2.0

\* Note: The lower beam, 15 feet thick, was shown to fail. This thickness is in the second beam above the mined seam.



This table shows that massive roof collapse is not likely if bedding planes or discontinuities are farther apart than the minimum beam thickness. Our observations of the core indicates bedding planes and discontinuities are generally greater than 1 to 1.5 feet apart. As a result we would not predict massive roof collapse to be a common failure mode. The only analyzed location where the calculation predicted roof collapse was within the first 15 feet above the No. 5 Seam. That prediction appears to be reasonably consistent with conditions encountered in the borings.

### C. Pillar Crushing

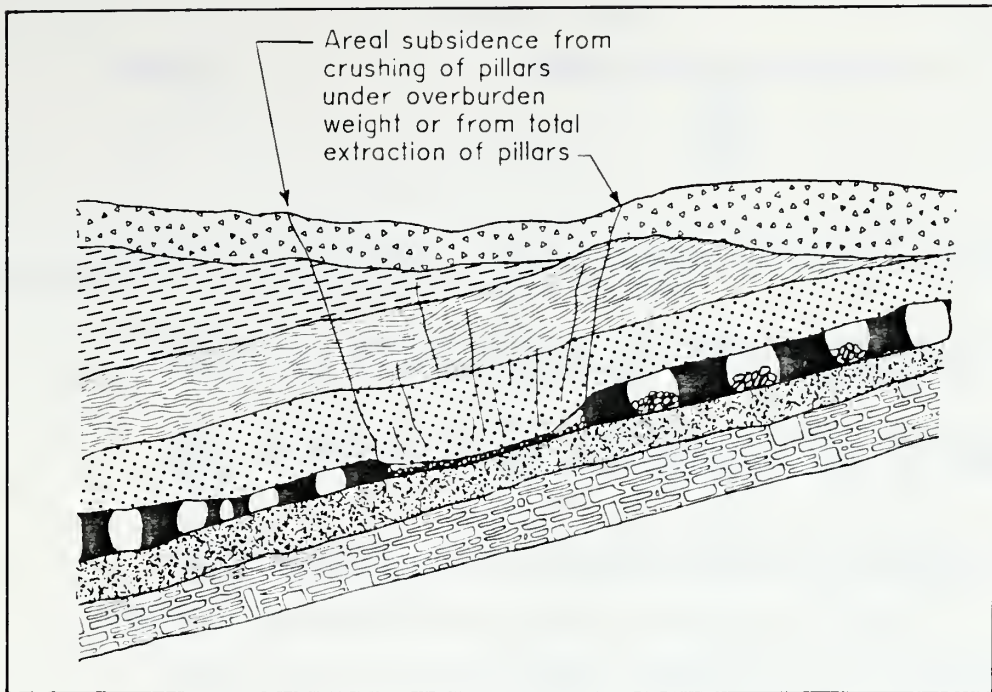
Pillars crush when the overburden pressure exceeds the strength of the pillars. Once a pillar fails the overburden load it carried is either transferred to adjacent pillars, which may in turn fail, or the roof collapse mode of failure might occur. Figure 5 illustrates the concept of tributary areas for calculating loads supported by pillars.

As noted earlier, the coal strength has been estimated using cubes cut from samples originating from the Smith No. 2 Seam. Sample calculations and computer generated results for this mode are presented in Appendix B. The analysis included correction factors for the size of cubes and pillar height to width ratios. The calculation is most sensitive to overburden pressure and strength of the coal. As stated previously we believe the critical load condition occurred during mining when rooms were dewatered. The flooded condition generally reduces the overburden pressure and thus increases the safety factor by a factor of about 1.7.

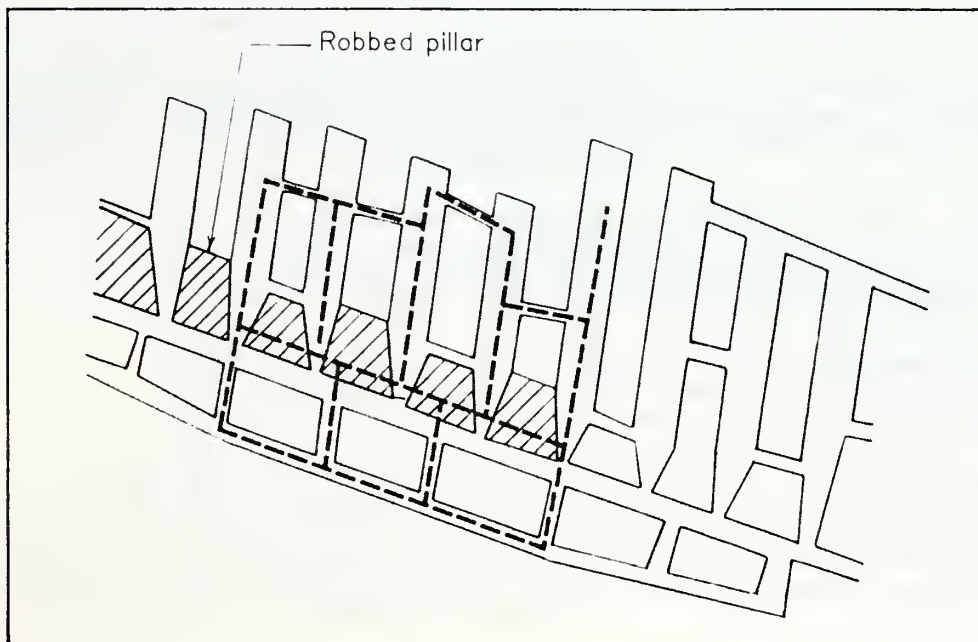
The condition of dipping beds creates changing overburden pressures along the length of rooms. For simplicity we used the average overburden pressure throughout each area analyzed. It is not known how the dipping beds influence the state of stress within pillars. It can be argued that a portion of the overburden load is transmitted through pillars in shear rather than purely compression. Considering that the long dimension of pillars is generally in the direction of dip and the length to height ratio is usually greater than 5 to 10, we believe the influence of these dip angles is negligible.

The rock profiles analyzed were typical of DH-1 and 3. Room and pillar configurations were selected which appeared representative. The calculations generally yield safety factors in the range of 2 to 3 for the submerged condition. Predictably safety factors will decrease down dip, and in deeper seams. For the dewatered condition which existed during mining, pillar failures are often predicted. It is possible the laboratory tests under estimate the coal strength in those seams, or stress redistributions (temporary arching) influenced performance during mining. Observation of the mine maps do not normally show pillar sizes increasing at greater depths. The following table presents the approximate depths below which we would predict pillar crushing to occur for the submerged condition.





5a. Cross Section of Loaded Pillars



5b. Plan View of Typical Pillar Configurations

Figure 5 Pillar Crushing Failure Mode



<u>Seam No.</u>	<u>Approximate Crushing Depth, ft.</u>
1-1/2	700
2	700
4	350
5	800

Comparing these depth estimates with the cross sections on Drawing No. 87-3001.D-7, the following observations are made:

- On the east bench cross section all mines extend significantly deeper than our calculations would predict possible.
- On the Main Street cross section, the mines generally stop near the depths we predict can be supported for the submerged case.
- On the airport cross section, the No. 1-1.2 and No. 4 seams extend below the depths our calculations predict possible.

From these observations the following conclusions are made:

1. Either the coal has greater strength than predicted by laboratory tests, or pillars in these deeper sections have been crushed.
2. Since the mines have been inactive for at least 50 years and flooding has reduced overburden pressures to buoyant conditions, future pillar crushing is unlikely.

#### D. Pillar Punching

Analysis methods for pillar punching have been developed (8 and 9) which are considered reasonably accurate. Based upon conditions of materials encountered in the borings however, we do not consider a calculation for pillar punching necessary for the following reasons.

- The thickness of soft underclay encountered in the core was thin, typically about one foot or less.
- Floor materials beneath the underclay are very competent such that pillars would obviously crush before the floor would yield.
- The consistency of the underclay is so soft that it was so badly disturbed by coring that samples for strength tests could not be obtained. It is the weakest material encountered.

From these observations we conclude that pillars have probably already punched some distance, but the underlying competent materials prevented punching of more than a few feet. We perceive this limited depth of punching as being the potential catalyst to initiating the limited roof strain and collapse. As pillars move down roof support decreases, and in effect increases the distance roof beams must span.





## E. Trough Subsidence Geometries

These calculations were performed utilizing methods presented and referenced by Peng (6). The technique was developed for use in British longwall mines and is generally considered to over estimate subsidence in U.S. mines, especially room and pillar mines. It does allow some modification to trough geometry due to dipping beds. This is purely a graphical method which ignores material strength properties except for their influence on the angle of draw.

Figures 6, 7, 8 and 9 present typical trough geometries for the following cases.

- No. 1 1/2 Seam for subsidence of a single room.
- No. 4 Seam for subsidence of two adjacent rooms if pillar is robbed.
- No. 4 Seam for subsidence of numerous adjoining rooms to simulate mass pillar robbing.
- No. 4 and 5 Seams for subsidence of two overlying areas of mass pillar robbing.

The last case of two overlying areas is not presented in the literature. Our analysis was to predict the trough geometry for subsidence of each area and then use superposition to construct the final geometry.

Figures 6 through 9 suggest that surface subsidence from a single room might only be in the range of a few inches, while multiple room and multiple seam subsidence is predicted to result in much greater surface subsidence. These figures are presented only for informational purposes. We are not aware of any previous subsidence features which can be represented by these constructions. Furthermore the condition of mine roofs and rooms observed in the borings present no indication which would lead us to believe the figures represent potential future surface subsidence. All indications suggest that bulking is preventing mine room failures from being expressed on the surface, except where chimneys have developed in high potential areas.

## V. SUMMARY AND CLOSING

Through the process of reviewing mine maps, strategic location of exploration borings, observing subsidence features and performing calculations using representative material properties we believe the following summary statements can be made.

1. Evidence of rock strain is generally limited to less than 20 feet above predicted mine roofs.
2. The perceived failure mode consists of a limited distance of pillar punching followed by some roof collapse which has probably stabilized.
3. Because most of the mines are now flooded, present day conditions are generally less conducive to subsidence than existed during or shortly after the mines were operating.



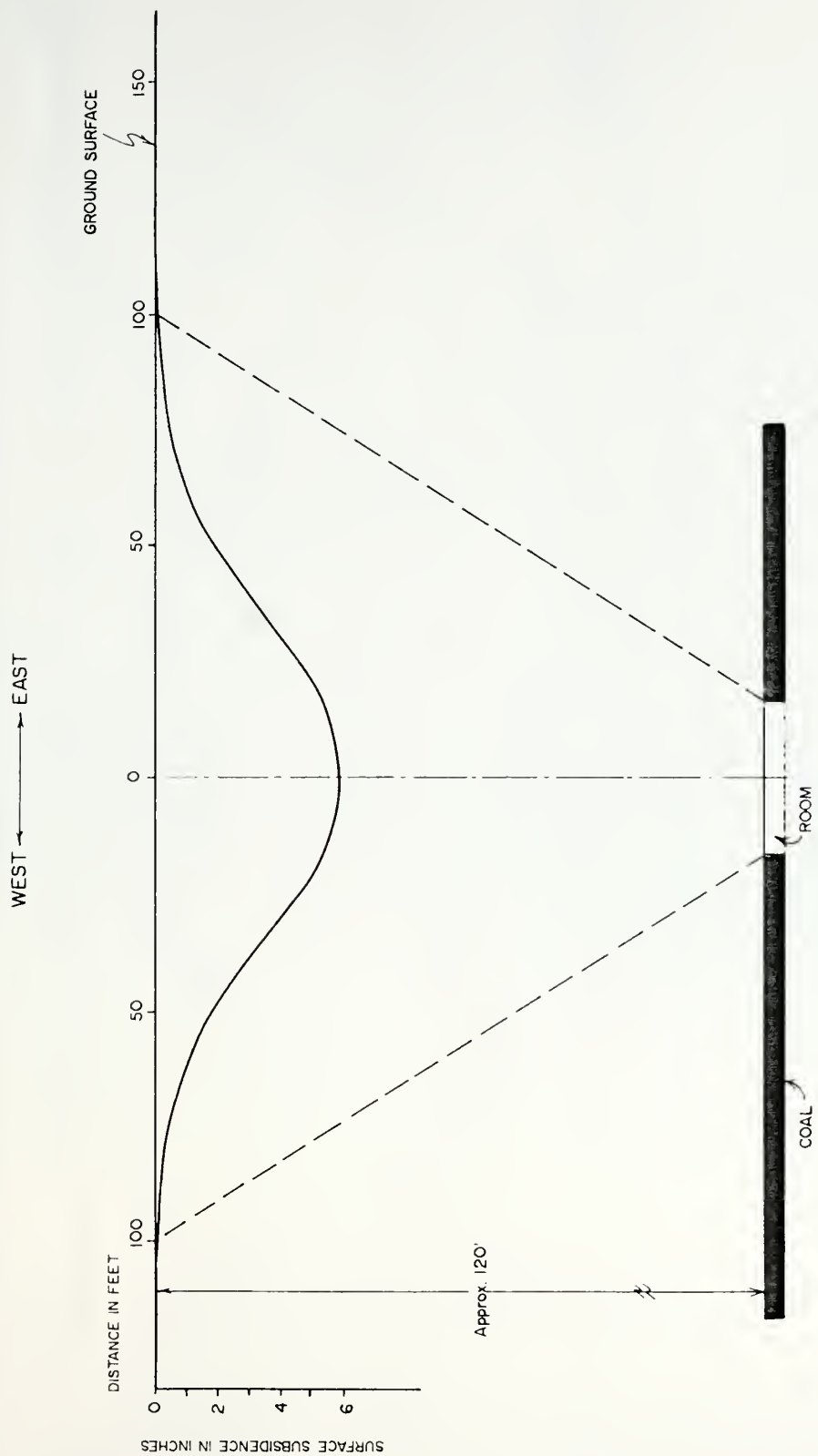


Figure 6 East-West Trough Geometry for Single Room in No. 1-1/2 Seam



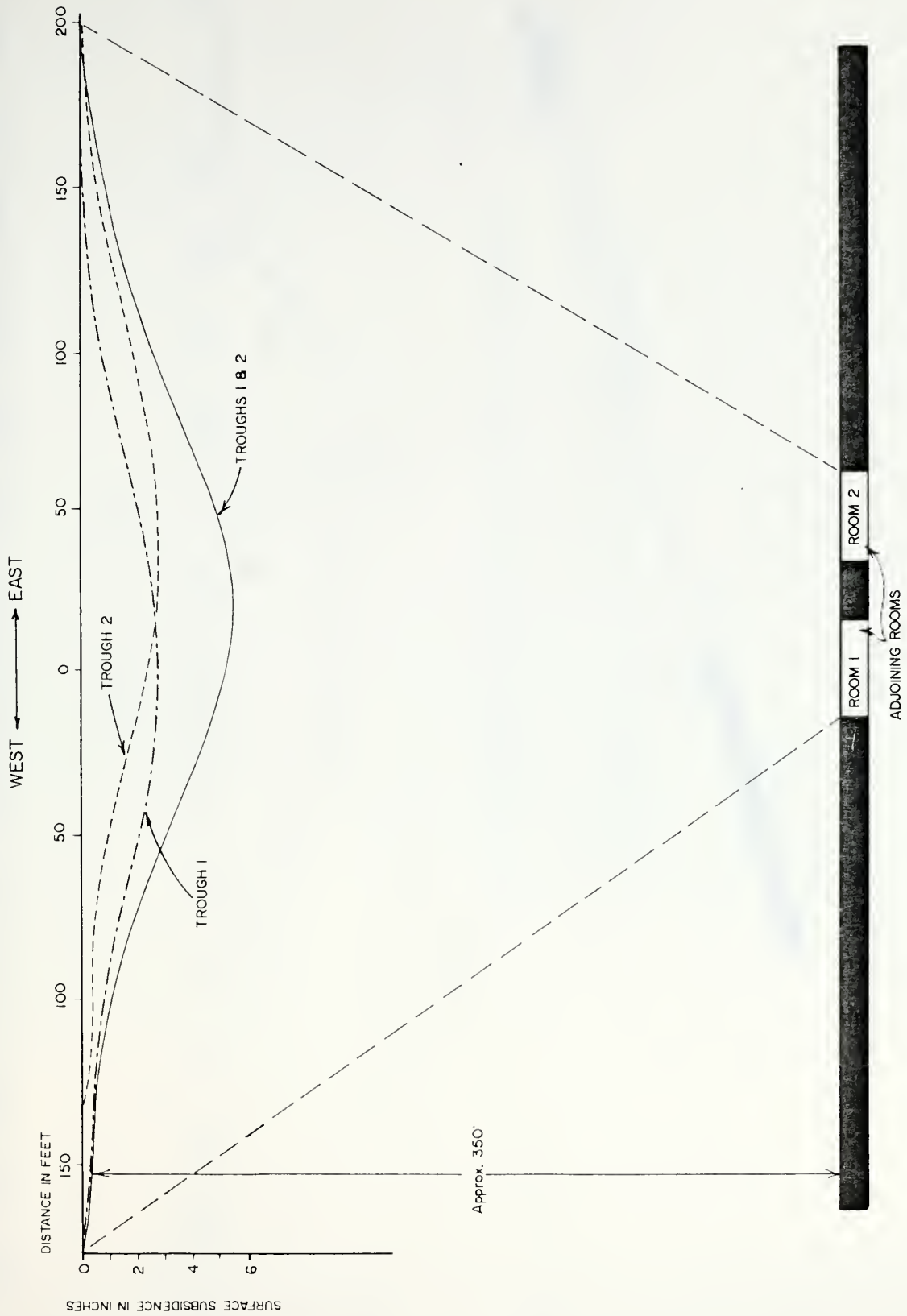


Figure 7 East-West Trough Geometry for 2 Adjoining Rooms in No. 4 Seam



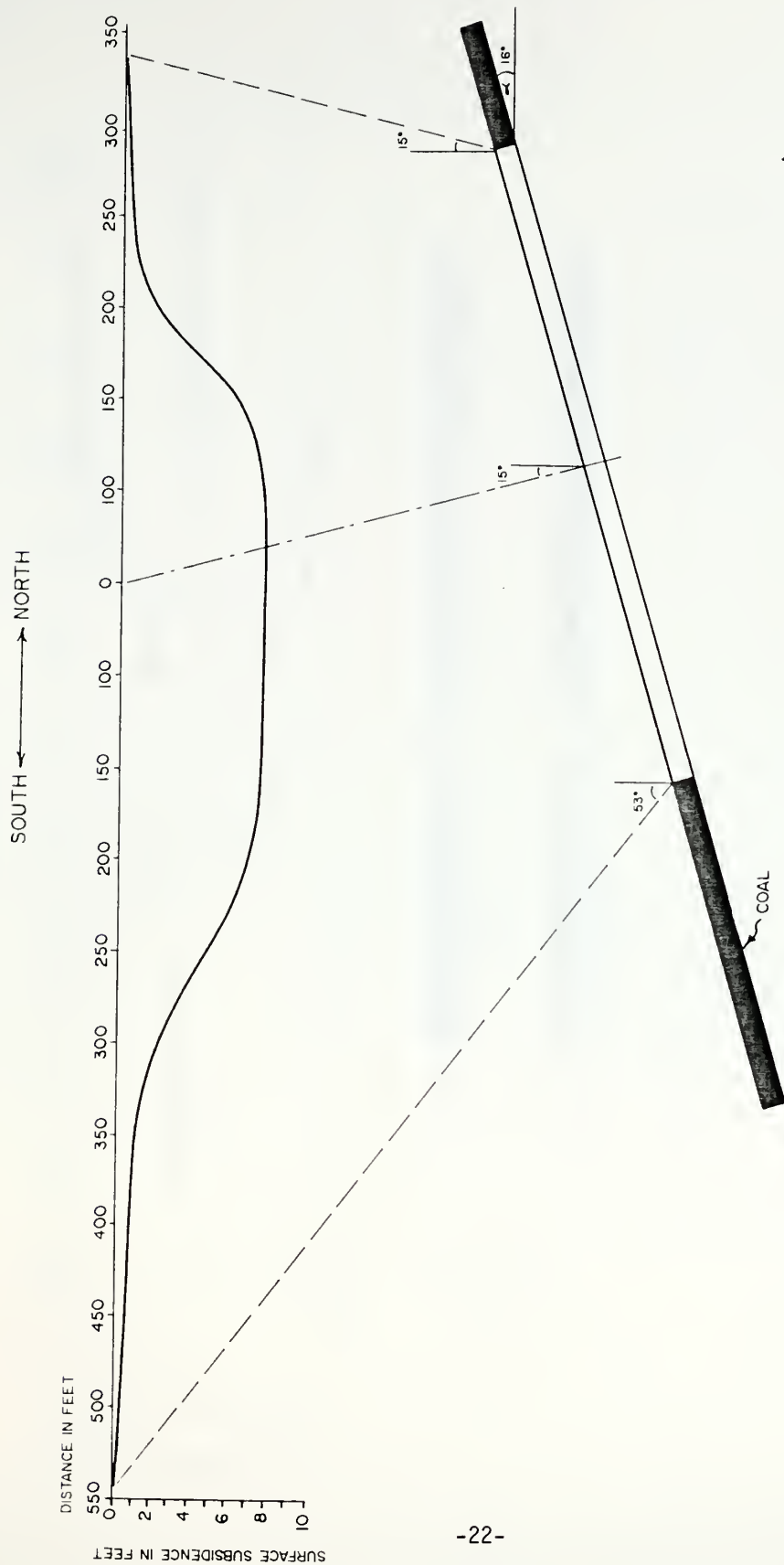


Figure 8 North-South Trough Geometry for Multiple Room Extractions in No. 4 Seam





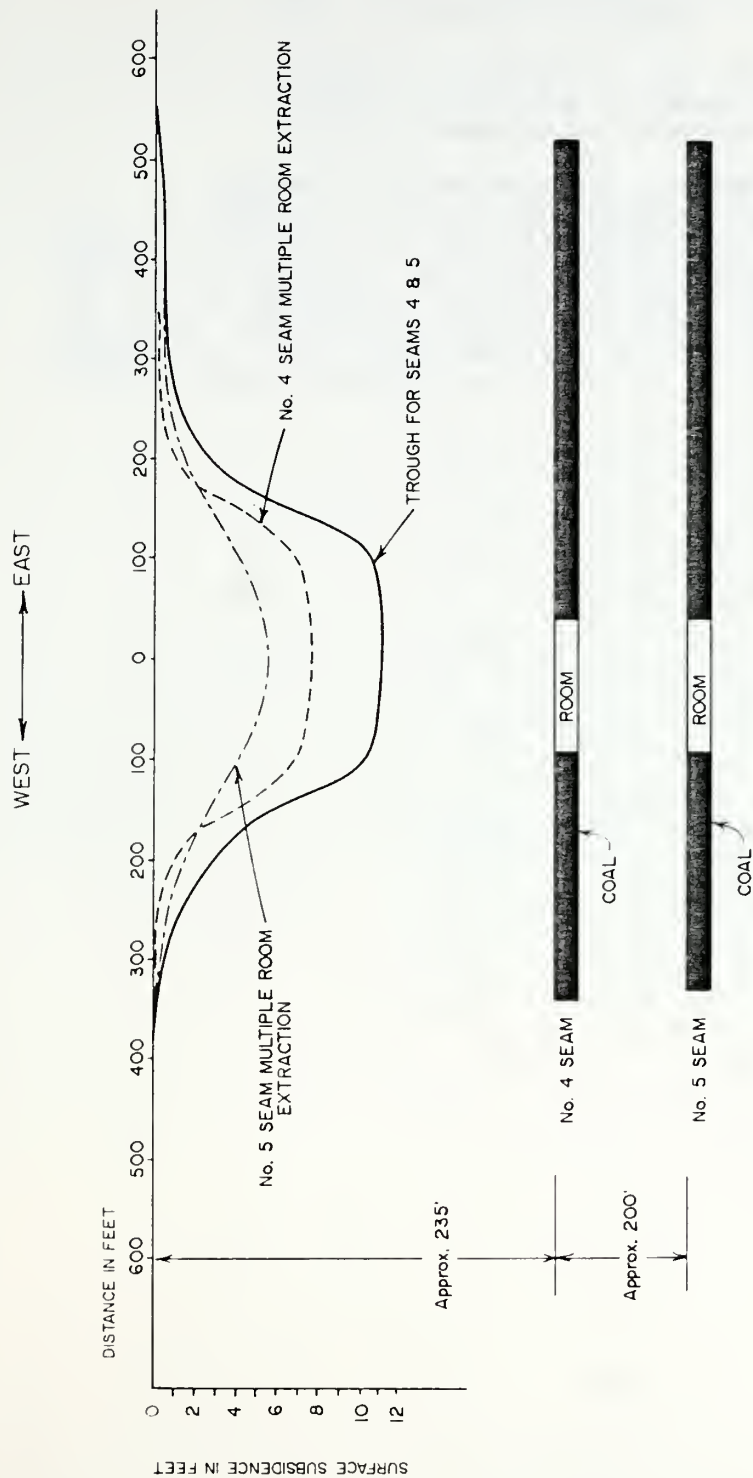


Figure 9 East-West Trough Geometry for Multiple Room Extractions in Seams 4 & 5



4. The potential for future subsidence is considered high only on the Bill Palmer property east of Red Lodge, near the west end of 10th Street, and above shallow adits in the Bearcreek area. It is predicted to consist of chimney subsidences similar to previous occurrences.
5. Two small areas within the town and part of the runway classify as having a moderate potential. The Red Lodge Subsidence Potential map is presented on page 12.

In making the above statements we concede the fact that the mines cover an extensive area and our subsurface explorations are very limited in relation to the mined area. It is therefore possible that conditions different than those encountered exist and might produce future subsidence. However, in view of the information contained in this report and the historical development of subsidence in other western coal fields we believe the potential for future surface expression of mine subsidence beneath presently developed areas is generally low.



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## APPENDIX A

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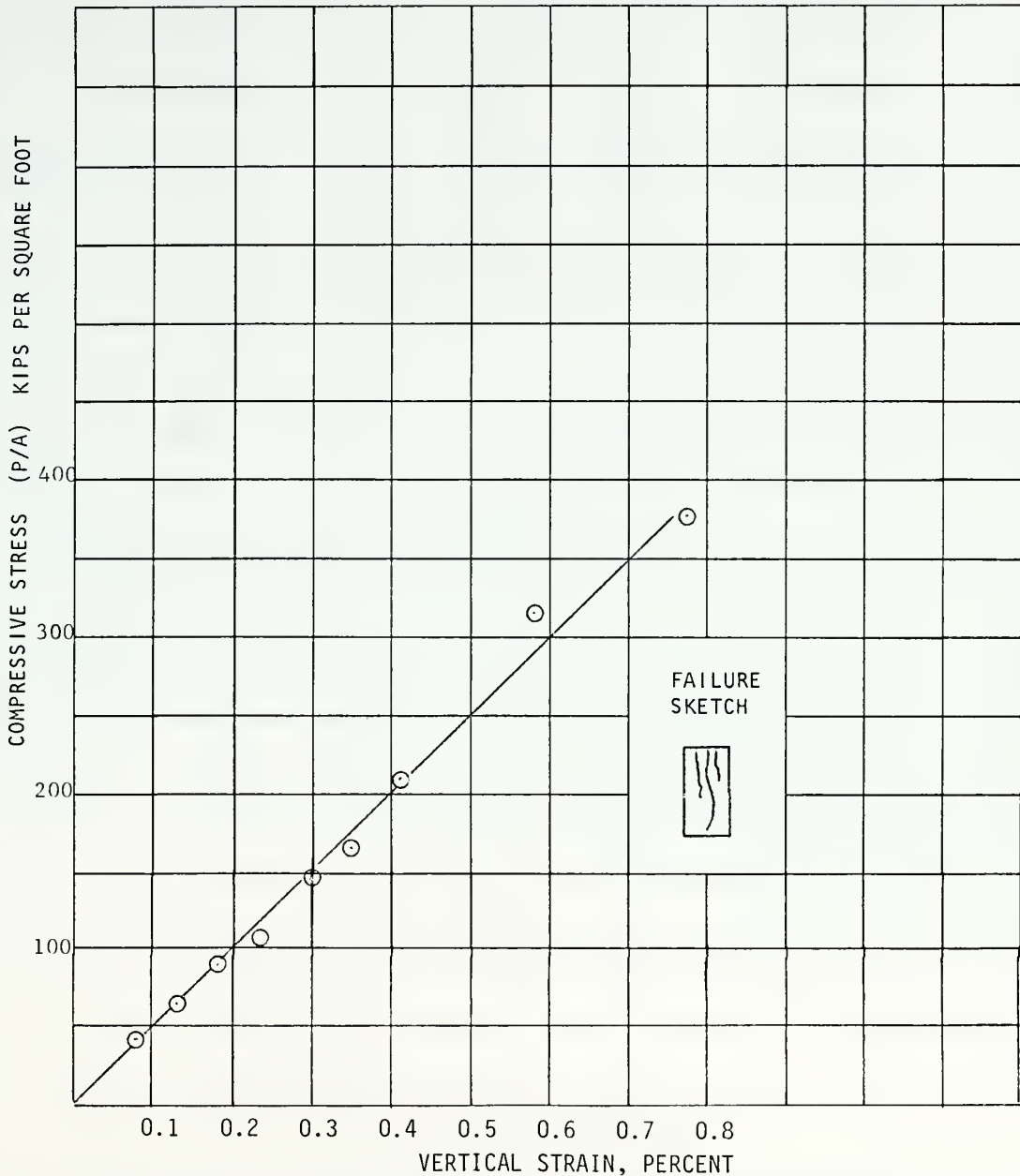




# UNCONFINED COMPRESSION TEST

DRILL HOLE DH-1  
DEPTH 135.0'-136.0'  
SAMPLE NO.

MOIST UNIT WEIGHT: 153 pcf.  
DRY UNIT WEIGHT : 146 pcf.  
MOISTURE CONTENT : 4 %  
CLASSIFICATION : Sandstone  
HEIGHT TO DIAMETER RATIO: 2.06  
RATE OF STRAIN: 0.08%/min.



RED LODGE/BEARCREEK SUBSIDENCE STUDY  
RED LODGE, MONTANA

STATE OF MONTANA DEPARTMENT OF STATE LANDS  
ABANDONED MINE RECLAMATION BUREAU  
HELENA, MONTANA



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and Testing, Inc.

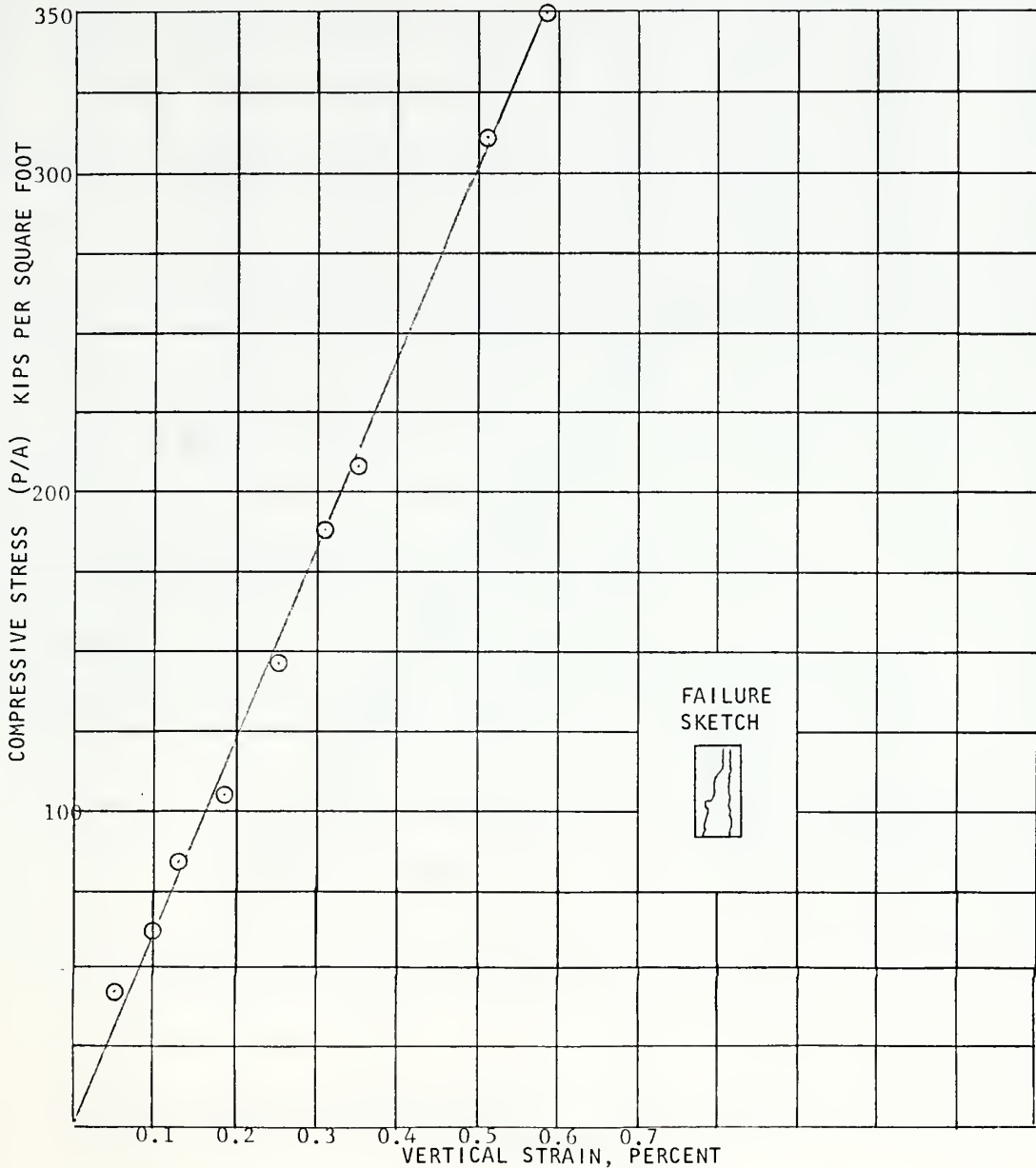
JOB NO. 87-3001.D PLATE NO. 1



# UNCONFINED COMPRESSION TEST

DRILL HOLE DH-1  
DEPTH 136.0'-137.0'  
SAMPLE NO.

MOIST UNIT WEIGHT: 153 pcf.  
DRY UNIT WEIGHT : 147 pcf.  
MOISTURE CONTENT : 4 %  
CLASSIFICATION : Sandstone  
HEIGHT TO DIAMETER RATIO: 2.05  
RATE OF STRAIN: 0.10%/min.



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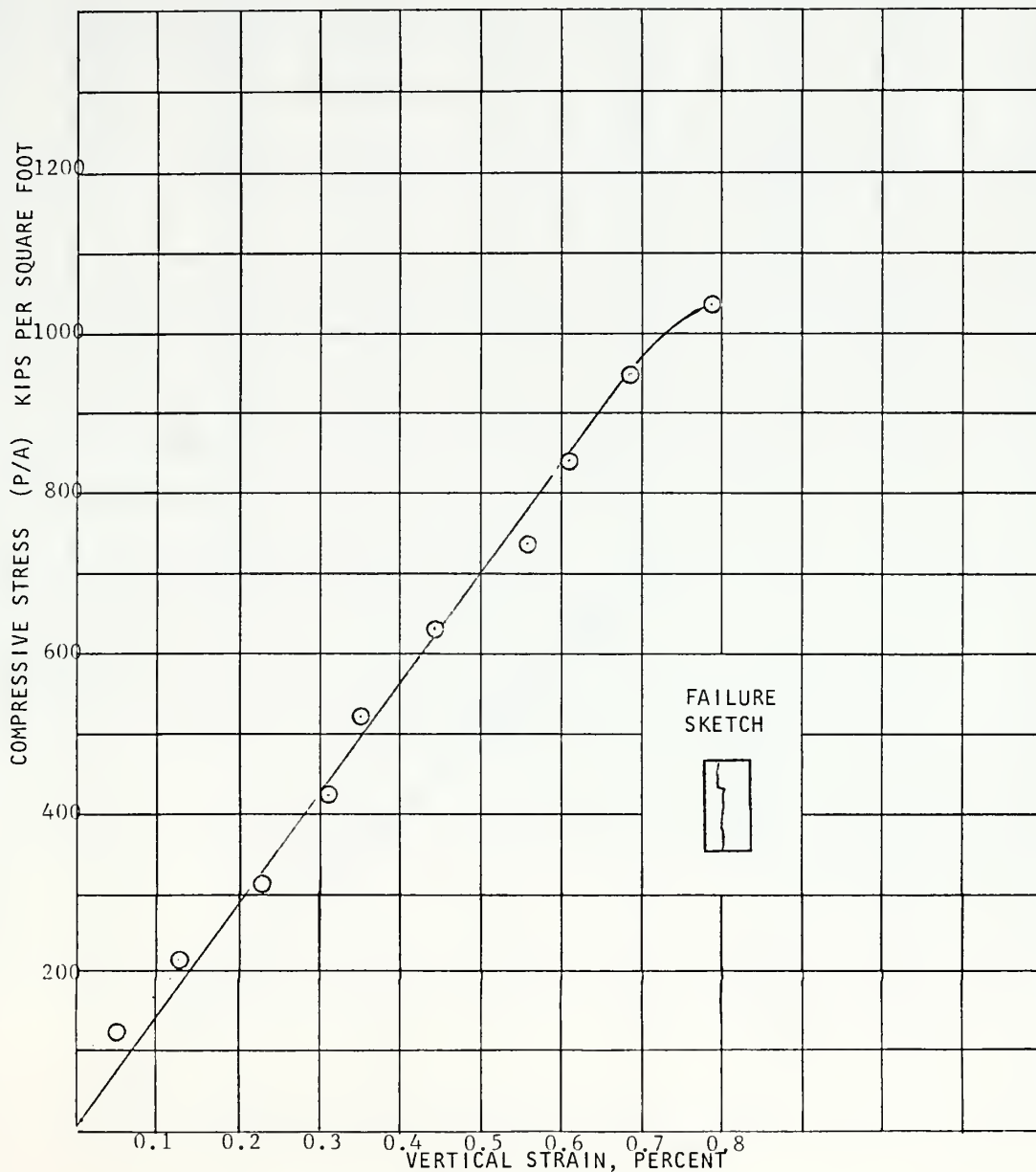
JOB NO. 87-3001.D PLATE NO. 2



## UNCONFINED COMPRESSION TEST

DRILL HOLE DH-3  
DEPTH 212.0'-213.0'  
SAMPLE NO.

MOIST UNIT WEIGHT: 155 pcf.  
DRY UNIT WEIGHT : 152 pcf.  
MOISTURE CONTENT, : 2 %  
CLASSIFICATION : Claystone  
HEIGHT TO DIAMETER RATIO: 2.05  
RATE OF STRAIN: 0.09%/min.



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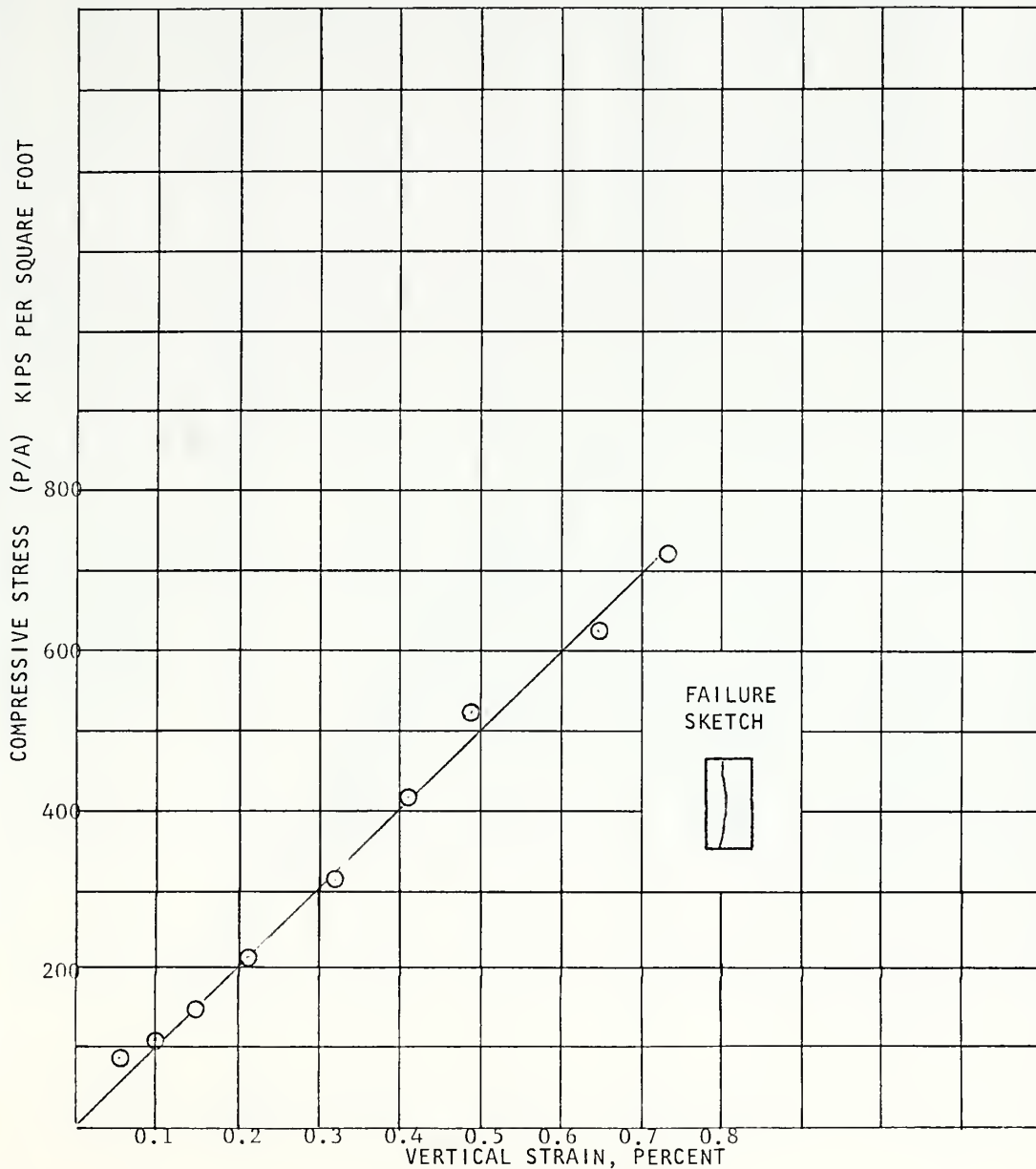
JOB NO. 87-3001.D PLATE NO. 3



# UNCONFINED COMPRESSION TEST

DRILL HOLE DH-4  
DEPTH 225.4'-226.3'  
SAMPLE NO.

MOIST UNIT WEIGHT: 155 pcf.  
DRY UNIT WEIGHT : 151 pcf.  
MOISTURE CONTENT : 2 %  
CLASSIFICATION : Claystone  
HEIGHT TO DIAMETER RATIO: 2.09  
RATE OF STRAIN: 0.06%/min.



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JOB NO. 87-3001.D PLATE NO. 4

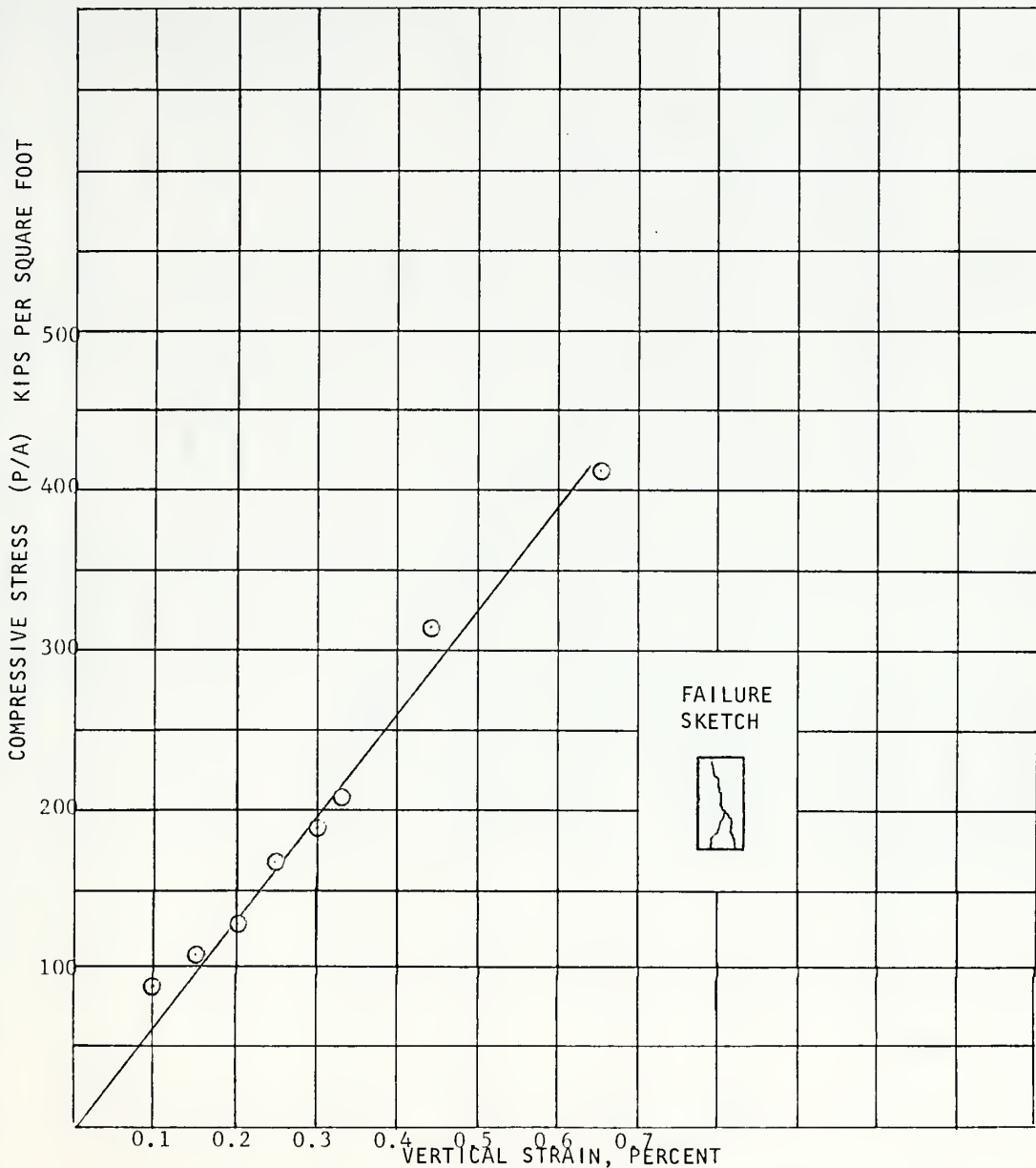




# UNCONFINED COMPRESSION TEST

DRILL HOLE DH-3  
DEPTH 414.0'-414.7'  
SAMPLE NO.

MOIST UNIT WEIGHT: 145 pcf.  
DRY UNIT WEIGHT : 138 pcf.  
MOISTURE CONTENT, : 5 %  
CLASSIFICATION : Sandstone  
HEIGHT TO DIAMETER RATIO: 2.06  
RATE OF STRAIN: 0.09%/min.



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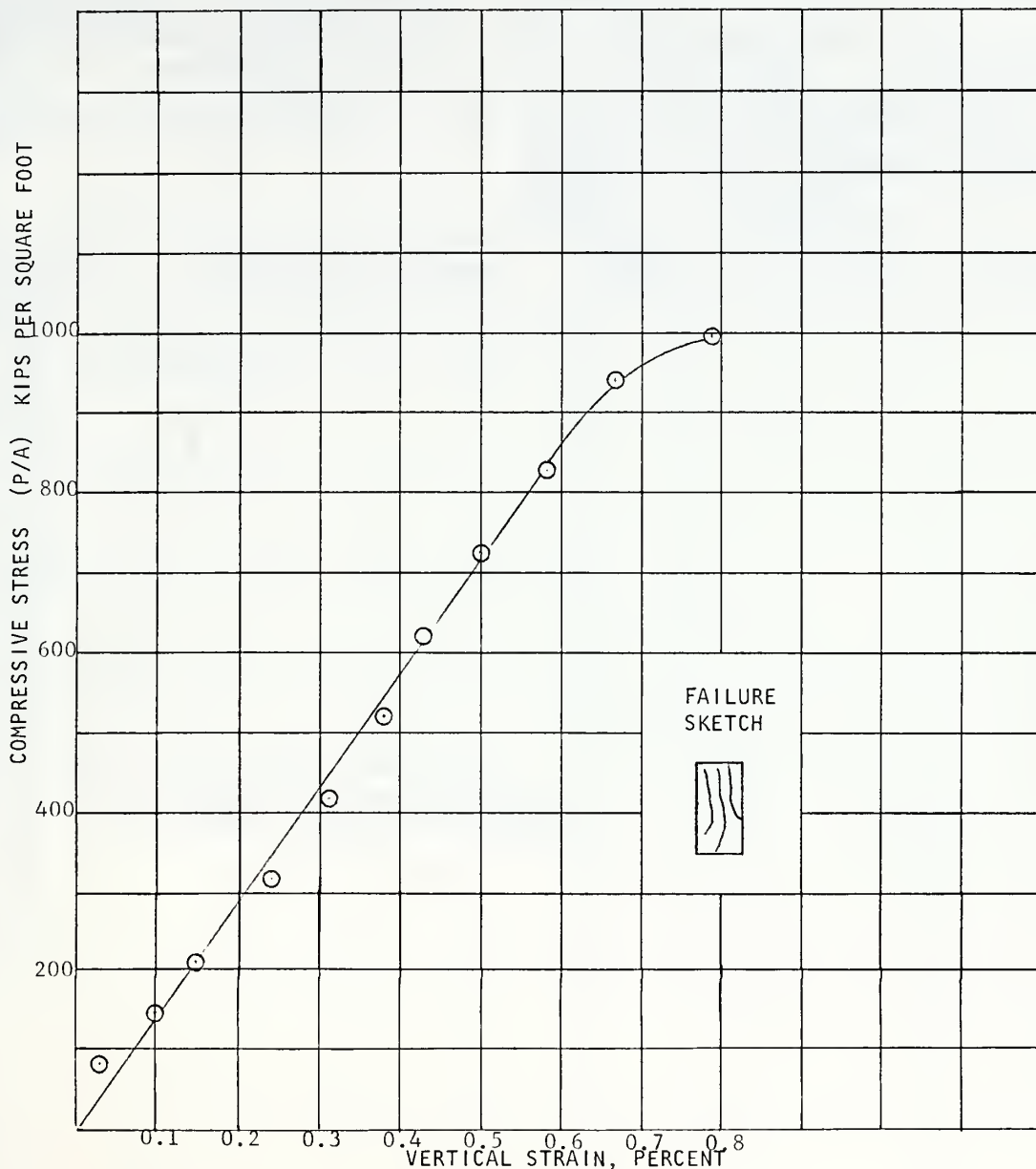
JOB NO. 87-3001.D PLATE NO. 5



# UNCONFINED COMPRESSION TEST

DRILL HOLE DH-3  
DEPTH 441.2'-442.0'  
SAMPLE NO.

MOIST UNIT WEIGHT: 154 pcf.  
DRY UNIT WEIGHT : 151 pcf.  
MOISTURE CONTENT : 2 %  
CLASSIFICATION : Sandstone  
HEIGHT TO DIAMETER RATIO: 1.97  
RATE OF STRAIN: 0.13%/min.



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JOB NO. 98-3001.D PLATE NO. 6





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## LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-1

SHEET 1 OF 3

DRILL TYPE: SOIL ROCK Schramm Rotadrill		CLIENT State of Montana AMR Program	
SIZE, TYPE OF BIT 3" Core 7-7/8 & 6-1/4 Tricone		PROJECT Red Lodge/Bearcreek Subsidence Study	
CASING: SIZE 8-5/8" LENGTH 108'		LOCATION Red Lodge Airport	
TOTAL NO. OF OVERBURDEN SAMPLES TAKEN		ELEVATION: TOP OF HOLE Not Determined	
Disturbed	Undisturbed	GROUNDWATER 118'	
TOTAL NO. CORE BOXES 5	TOTAL CORE RECOVERY FOR BORING (%) 77	THICKNESS OF OVER-BURDEN, FT. 91.0	DEPTH DRILLED IN-TO ROCK, FT. 134
REMARKS		TOTAL DEPTH OF HOLE, FT. 225	
		STARTED 7-20-87	COMPLETED 7-23-87
		DRILLED BY B. Kupfner	
		LOGGED BY R. Dombrowski	

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOVERY	R.Q.D.	DEPTH, FEET	BLOWS/FOOT
7.0		FILL; Mine Spoils				
20						
40		Poorly Graded GRAVEL with Cobbles and Boulders; very dense, rounded to subrounded, difficult penetration.				
60						
80		(continued .....)				



# LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-1

SHEET 2 OF 3

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOVERY	R Q D	DEPTH, FEET	GEOMECHANICAL			
						DEGREE OF WEATHERING	JOINTS/FT	DISCONTINUITES	
								TYPE	DIP
80		Poorly Graded GRAVEL with Cobbles and Boulders.							
91		Lean CLAY; very stiff, brown, decomposed claystone.							
101		SANDSTONE; gray, very fine grained, moderately hard to hard rock.							
108		CLAYSTONE; light to dark gray, soft to moderately hard rock, interbedded with thin sandstone seams.							
118		GWL (7-23-87)							
		Becoming broken with thin coal and carbonaceous seams below 117 feet.				SW	I	BJ OX	17°
129		COAL No. 1-1/2 Seam, highly fractured.	50	21	120-134	SW	I	OJ OX	85°
134		SANDSTONE; gray, very fine grained, moderately hard to hard rock.				SW	I	BJ	80°
146.7		CLAYSTONE; dark gray.							
150		SANDSTONE; light gray, very fine grained, argillaceous, moderately hard rock.	96	87	134-154	SW	I	CJ	20°
160									
180									
190		(continued .....)							





# LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-1

SHEET 3 OF 3

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOVERY	R Q D	DEPTH, FEET	BLOWS/FOOT
190	...	SANDSTONE; light gray, very fine grained, argillaceous, moderately hard rock.				
200		VOID				
205	0 0 Δ	Broken Rock Rubble				
216	Δ Δ Δ	CLAYSTONE; Mine Floor				
225		Bottom of Hole				





































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LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH - 2

SHEET 1 OF 4

DRILL TYPE: SOIL Bucyrus Erie  
ROCK

CLIENT

State of Montana AMR Program

SIZE, TYPE OF BIT  
3" Core, 7-7/8 Tricone

PROJECT

Red Lodge/Bearcreek

CASING: SIZE 8-5/8 LENGTH 75'

LOCATION East Bench

TOTAL NO. OF OVERBURDEN SAMPLES TAKEN

8' North of section corner 7,26,4,35

Disturbed Undisturbed

ELEVATION: TOP OF HOLE

GROUNDWATER

TOTAL NO. TOTAL CORE RECOVERY

Approximately 5835

27.3

CORE BOXES 8 FOR BORING (%) 97

THICKNESS OF OVER- BURDEN, FT. 21.0 DEPTH DRILLED IN- TO ROCK, FT. 349 TOTAL DEPTH OF HOLE, FT. 370'

REMARKS

STARTED 8-10-87 COMPLETED 8-19-87

DRILLED BY Rock Creek Drilling (B. Kupfner)

LOGGED BY R. Dombrowski

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOV- ERY	R.Q.D.	DEPTH, FEET	BLOWS/FOOT
0.4		TOPSOIL				
		Lean CLAY; firm, moist to wet, occasional sand seams.				
21						
27.3		GWL (8-19-87)				
40						
		CLAYSTONE; light gray to light brown, soft rock, thinly laminated, carbonaceous seams.				
60						
80						

(continued .....)



# LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-2

SHEET 2 OF 4

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOV- ERY	R Q D	DEPTH, FEET	BLOWS/FOOT
80						
87		CLAYSTONE				
97		SANDSTONE; Gray, very fine grained, soft rock, argillaceous.				
120		CLAYSTONE; Dark gray, thinly bedded, soft rock.				
135		SANDSTONE; Gray, very fine grained, soft rock, argillaceous.				
144		CLAYSTONE; Dark gray, thinly bedded, soft rock, thin carbonaceous layers.				
155		SANDSTONE; Light gray, soft rock.				
159		CLAYSTONE; Dark gray, soft to moderately hard rock, thinly laminated occasional coal and carbonaceous seams				
180						
		(continued .....				



# LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-2

SHEET 3 OF 4

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOV- ERY	R Q D	DEPTH, FEET	BLOWS/FOOT
190						
200						
220						
240		CLAYSTONE; Dark gray, soft to moderately hard rock, thinly laminated, occasional coal and carbonaceous seams.				
260						
280						
300		(continued .....)				



# LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-2

SHEET 4 OF 4

DEPTH, FEET	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOVERY	R Q D	DEPTH, FEET	GEOMECHANICAL			
						DEGREE OF WEATHERING	JOINTS / FT	DISCONTINUITIES	
								TYPE	DIP
300		CLAYSTONE; Dark gray, soft to moderately hard rock, thinly laminated, occasional coal and carbonaceous seams.							
310			100	100	310-313	SW	2	CJ	11°
		SANDSTONE; Light gray, very fine grained, moderately hard rock, argillaceous seams, some carbon detrius and thin coal layers.				SW		CJ	8°
						F	2	BJ	
320			100	96	313-333	SW	2	CJ	46°/88°
						SW	1	CJ	64°
328.7									
		CLAYSTONE; Dark gray, carbonaceous, sandy, moderately hard rock, thin sandstone seams, some fractured zones.	100	89	333-346.2	MW	2	CJ Gauge	53°
						MW	2	CJ	76°
						SW	1	CJ	65°
349		SANDSTONE; Light gray, very fine grained, hard rock.	89	81	346.2-360	SW	2	CJ	60°
355		CLAYSTONE; Gray, sandy, moderately hard rock				SW	1	CJ	89°
358		COAL; Highly fractured.				MW	1	CJ	
358.5									
		CLAYSTONE; Gray, soft at contact then hard.							
370		Bottom of Hole							





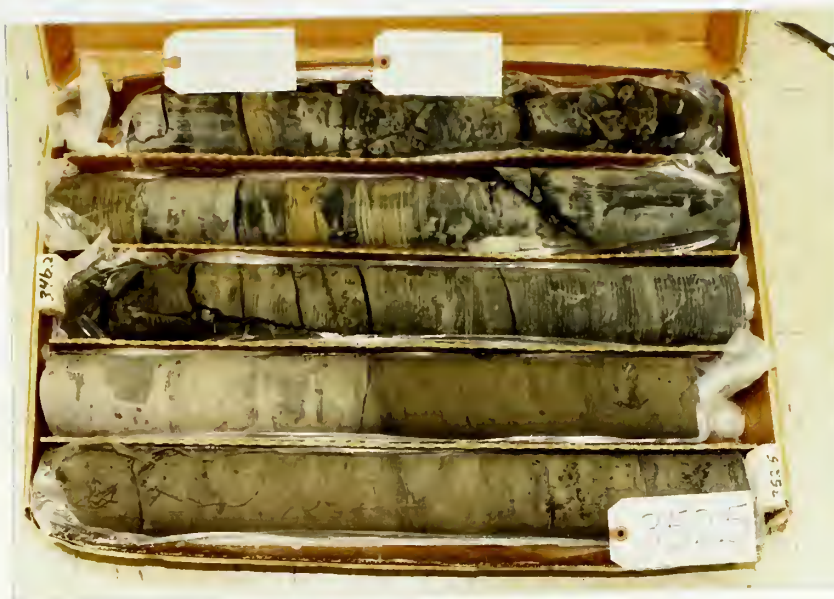
















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LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-3  
SHEET 1 OF 5

DRILL TYPE: SOIL		CLIENT	
ROCK Schramm Rotadrill		State of Montana AMR Program	
SIZE, TYPE OF BIT 7-7/8", 6-1/4" & 3" Core Tricones 10-5/8"		PROJECT Red Lodge/Bearcreek	
CASING: SIZE 8-6/8" LENGTH 110'		LOCATION On 14th Street, 170' West of Main Street, 10' South of North Curb	
TOTAL NO. OF OVERBURDEN SAMPLES TAKEN		ELEVATION: TOP OF HOLE GROUNDWATER	
Disturbed Undisturbed		Not Determined 27	
TOTAL NO.	TOTAL CORE RECOVERY	THICKNESS OF OVER- DEPTH DRILLED IN- TOTAL DEPTH OF	
CORE BOXES 9	FOR BORING (%) 76	BURDEN, FT. 108'	TO ROCK, FT. 342.5 HOLE, FT. 450.5
REMARKS		STARTED 7-23-87 COMPLETED 7-31-87	
		DRILLED BY Rock Creek Drilling	
		LOGGED BY R. Dombrowski	

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOVERY	R.Q.D.	DEPTH, FEET	BLOWS/FOOT
1.0		FILL; Silty Sand				
20						
27		GWL (7-31-87)				
40		Poorly Graded GRAVEL with cobbles and boulders; very dense, rounded to subrounded, difficult penetration.				
60						
80		(continued.....)				





# LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-3

SHEET 2 OF 5

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOVERY	R Q D	DEPTH, FEET	BLOWS/FOOT
80						
90						
100		Poorly Graded GRAVEL with cobbles and boulders; very dense, rounded to subrounded, difficult penetration.				
108						
120						
140		SANDSTONE; Gray, very fine grained, moderately well cemented, argillaceous.				
160						
180						
(continued.....)						



# LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-3

SHEET 3 OF 5

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOV- ERY	R Q D	DEPTH, FEET	GEOMECHANICAL			
						DEGREE OF WEATHERING	JOINTS/FT	DISCONTINUITES	
								TYPE	DIP
190		SANDSTONE; Gray, very fine grained, moderately well cemented, argillaceous.							
200									
205									
		CLAYSTONE; dark gray, sandy, moderately hard rock, laminated, becoming light gray and more sandy below 212 feet with some carbon detritus.							
						F	O	Bd	17°
						all breaks drilling induced			
220		COAL; Highly fractured	95	89	210-230				
227.8						Broken	Rock		
237		CLAYSTONE; Dark gray, carbonaceous, soft at contact, thinly laminated.	69	0	230-241				
245									
260									
280		SANDSTONE; Light gray, very fine grained, argillaceous.							
300		(continued .....)							



## LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-3

SHEET 4 OF 5

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOV- ERY	R Q D	DEPTH, FEET	BLOWS/FOOT
300						
320						
340		SANDSTONE; Light gray, very fine grained, argillaceous.				
360						
380						
400						
(continued .....)						



# LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

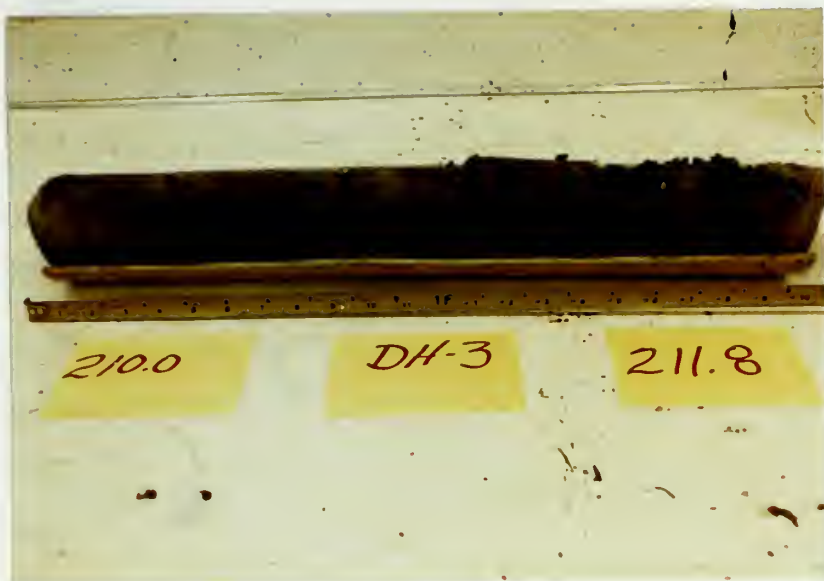
HOLE NO. DH-3

SHEET 5 OF 5

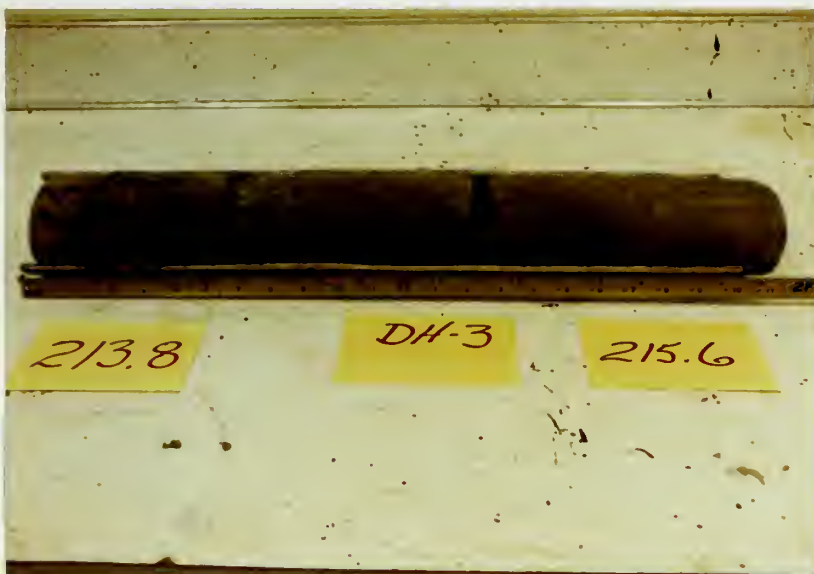
DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOVERY	R Q D	DEPTH, FEET	GEOMECHANICAL			
						DEGREE OF WEATHERING	JOINTS/FT	DISCONTINUITES	
								TYPE	DIP
410		SANDSTONE: Light gray, very fine grained, argillaceous, with thin coal and claystone seams.	99	86	410-429	SW	3	BJ	18°
420						SW	1	BJ	
						SW	1	BJ	
		CLAYSTONE; Dark gray, soft rock, thinly laminated, coal seams, slickensides, thin broken zones.	80	15	429-443	SW	2	QJ	10°
						MW	2	QJ	60°
						MW	1	CJ	50°
435		MISSING - Lost core	100	51	443-450.5	MW	9	QJ	52°
437						F	3	BJ	0°
						F	1	CJ	75°
		SANDSTONE; Gray, fine grained, thin claystone seams.							
444		CLAYSTONE: Gray, laminated, sandy layers.							
450.5		Bottom of Hole							

































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LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-5

SHEET 1 OF 3

DRILL TYPE: SOIL ROCK	Bucyrus Erie	CLIENT	State of Montana AMR Program	
SIZE, TYPE OF BIT	7-7/8" Tricone	PROJECT	Red Lodge/Bearcreek	
CASING: SIZE 7" LENGTH 57.5'		LOCATION	Washoe Sloped Entry North Side Highway 308	
TOTAL NO. OF OVERBURDEN SAMPLES TAKEN		ELEVATION: TOP OF HOLE	GROUNDWATER	
Disturbed		4979*	18.0'	
Undisturbed				
TOTAL NO. CORE BOXES N/A	TOTAL CORE RECOVERY FOR BORING (%) N/A	THICKNESS OF OVER- BURDEN, FT. 47.0'	DEPTH DRILLED IN- TO ROCK, FT. 204.0	TOTAL DEPTH OF HOLE, FT. 251.0
REMARKS		STARTED 7-14-87	COMPLETED 7-16-87	
		DRILLED BY B. Kupfner		
		LOGGED BY R. Dombrowski		

\*Elevation interpolated from  
topographic map.

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOV- ERY	R.Q.D.	DEPTH, FEET	BLOWS/FOOT
0.3		TOPSOIL				
18		GWL (7-16-87)				
45		Lean CLAY; Stiff, moist to saturated, dark brown.				
47		Clayey GRAVEL				
60		CLAYSTONE; Dark gray, soft rock, thinly bedded, occasional thin sandstone and coal layers.				
80		(continued.....)				





# LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-5

SHEET 2 OF 3

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOVERY	R Q D	DEPTH, FEET	BLOWS/FOOT
80		CLAYSTONE; dark gray, soft rock, thinly bedded, occasional thin sandstone and coal layers.				
100						
105						
125		SANDSTONE; Dark gray, fine grained, moderately cemented, argillaceous.				
140		CLAYSTONE; Dark gray, soft to moderately hard rock, thinly laminated, occasional thin sandstone seams.				
155						
160		SANDSTONE; Dark gray, fine grained, moderately cemented, argillaceous.				
169						
180		CLAYSTONE; Dark gray, moderately hard to hard rock, numerous thin sandstone and coal layers.				
		(continued.....)				





## LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-5

SHEET 3 OF 3

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOV- ERY	R Q D	DEPTH, FEET	BLOWS/FOOT
190						
200						
		CLAYSTONE; Dark gray, moderately hard to hard rock, numerous thin sandstone and coal layers.				
220						
237		COAL				
245		SANDSTONE; Dark gray, fine grained, argillaceous, moderately cemented.				
251		Bottom of Hole				





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LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-6

SHEET 1 OF 3

DRILL TYPE: SOIL ROCK Schramm Air Drill		CLIENT State of Montana AMR Program		
SIZE, TYPE OF BIT 7-7/8 Tricone		PROJECT Red Lodge/Bearcreek		
CASING: SIZE 7" LENGTH 46.7'		LOCATION Washoe Mine Highway 308 South of 4-Mile Marker		
TOTAL NO. OF OVERBURDEN SAMPLES TAKEN		ELEVATION: TOP OF HOLE GROUNDWATER		
Disturbed Undisturbed		Not Determined 19		
TOTAL NO.	TOTAL CORE RECOVERY	THICKNESS OF OVER-	DEPTH DRILLED IN-	TOTAL DEPTH OF
CORE BOXES	FOR BORING (%)	BURDEN, FT. 38'	TO ROCK, FT. 228	HOLE, FT. 226
REMARKS		STARTED 7/16/87	COMPLETED 7/16/87	
		DRILLED BY Bill Kupfner		
		LOGGED BY R. Dombrowski		

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOVERY	R.Q.D.	DEPTH, FEET	BLOWS/FOOT
0.3		TOPSOIL				
		Lean CLAY; stiff, moist to saturated, dark gray.				
19		GWL (7/16/87)				
31		Poorly Graded GRAVEL with Clay; dense, saturated.				
38		CLAYSTONE; dark gray, moderately hard rock, laminated, occasional sandstone layers.				
55		SANDSTONE; gray, hard rock, well cemented, very fine grained, argillaceous.				
80		(continued ....)				



# LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-6

SHEET 2 OF 3

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOVERY	R Q D	DEPTH, FEET	BLOWS/FOOT
80						
		SANDSTONE; gray, hard rock, well cemented, very fine grained, argillaceous.				
100						
120						
141		COAL				
144		SANDSTONE				
151		COAL				
154						
160						
		SANDSTONE; light gray, moderately hard rock, well cemented, fine grained, occasional claystone and coal layers.				
180						
		(continued ....)				



# LOG OF EXPLORATION BORING

JOB NO. 87-3001.D

HOLE NO. DH-6

SHEET 3 OF 3

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOV- ERY	R Q D	DEPTH, FEET	BLOWS/FOOT
190		CLAYSTONE; gray, moderately hard rock, thinly laminated, some carbon detritus.				
200						
215		SANDSTONE; light gray, moderately hard rock, well cemented.				
230		COAL				
236		SANDSTONE; gray, moderately well cemented, argillaceous.				
260						
266		Bottom of Hole				





**Northern**Engineering  
and Testing, Inc.**LOG OF EXPLORATION BORING**JOB NO. 87-3001.DHOLE NO. DH-7,7A,7BSHEET 1 OF 1

DRILL TYPE: SOIL ROCK Schramm Air Drill		CLIENT State of Montana AMR Program	
SIZE, TYPE OF BIT 6-1/4" Tricone		PROJECT Red Lodge/Bearcreek	
CASING: SIZE 7" LENGTH 7'		LOCATION Smith Mine; Mapped Entry west of entry into Red Lodge Coal Co.	
TOTAL NO. OF OVERBURDEN SAMPLES TAKEN		ELEVATION: TOP OF HOLE GROUNDWATER Not Encountered	
Disturbed	Undisturbed	THICKNESS OF OVER- DEPTH DRILLED IN- TOTAL DEPTH OF BURDEN, FT. TO ROCK, FT. HOLE, FT.	
TOTAL NO. CORE BOXES N/A	TOTAL CORE RECOVERY FOR BORING (%) N/A	STARTED 7/17/87 COMPLETED 7/17/87	
REMARKS		DRILLED BY Paul	
		LOGGED BY R. Dombrowski	

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOV- ERY	R.Q.D.	DEPTH, FEET	BLOWS/FOOT
0.5		TOPSOIL				
		FILL; Lean Clay; firm to stiff, moist, claystone fragments, brown.				
16		COAL				
22		SANDSTONE; light gray, hard rock, very fine grained.				
39		Bottom of Hole				

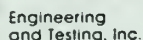


**Northern**Engineering  
and Testing, Inc.**LOG OF EXPLORATION BORING**JOB NO. 87-3001.DHOLE NO. DH-8SHEET 1 OF 1

DRILL TYPE: SOIL ROCK Schramm Rotadrill		CLIENT State of Montana AMR Program	
SIZE, TYPE OF BIT 6-1/4" Tricone		PROJECT Red Lodge/Bearcreek	
CASING: SIZE 7" LENGTH 7'		LOCATION South Highway 308; Unmapped Smith Mine Adit; E of Entry into Red Lodge Coal Co.	
TOTAL NO. OF OVERBURDEN SAMPLES TAKEN		ELEVATION: TOP OF HOLE GROUNDWATER	
Disturbed	Undisturbed	Not Determined 52	
TOTAL NO. CORE BOXES	TOTAL CORE RECOVERY FOR BORING (%)	THICKNESS OF OVER- BURDEN, FT.	DEPTH DRILLED IN- TO ROCK, FT.
REMARKS		TOTAL DEPTH OF HOLE, FT.	
		STARTED 7/17/87	COMPLETED 7/17/87
		DRILLED BY R. Dombrowski	
		LOGGED BY R. Dombrowski	

DEPTH, FT.	LEGEND	CLASSIFICATION OF MATERIALS	% CORE RECOV- ERY	R.Q.D.	DEPTH, FEET	BLOWS/ FOOT
2.0		Lean CLAY				
20		CLAYSTONE; brown to gray, soft rock, weakly cemented.				
40						
52		GWL (7/17/87)				
62		COAL				
68.5		CLAYSTONE				
70		Bottom of Hole				







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**APPENDIX B**

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PROJECT Rollidge / Boonville

JOB NO. 87-3001.D-5

PURPOSE Ref. Analysis: #1 1/2 room Crib

SHEET 1 OF 4

COMPUTED BY LT

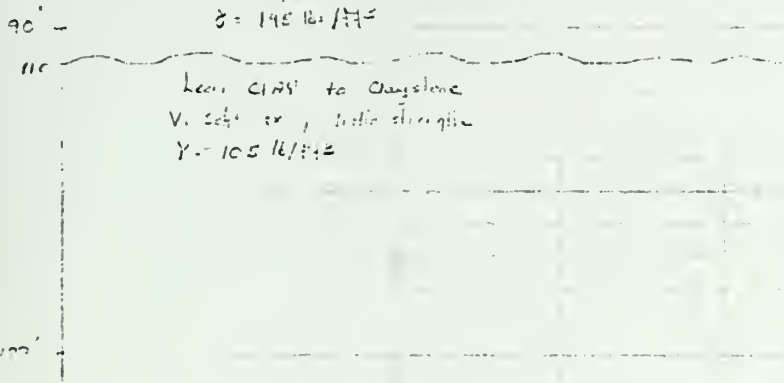
CHECKED BY JP

DATE 8-25-87

DM 1 V. thin interbedded sandstone and shales (immediate roof)

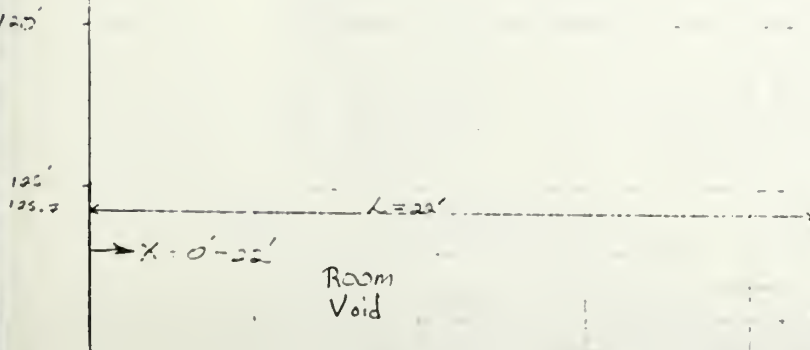
V. large Alluvial Gravel  
 $\gamma = 145 \text{ lb/ft}^3$

Comments



Irregularly stratified, Med Hard Gr  
 $\gamma = 150 \text{ lb/ft}^3$   
15' thick to 50' thick  
further stratified

V. thin interbedded shale w/ sandstone (Banded)  
15' thick, FMR - Poor R.R. (29 in dia)  
 $F_m = 5.65 \text{ ksi}$  ;  $8.455 \times 10^5 \text{ lb}$   
 $\gamma = 140 \text{ lb/ft}^3$  ;  $q_1 = 435 \text{ psi}$   
 $\phi = 30^\circ$   
 $C = 16,100 \text{ psi}$







PROJECT \_\_\_\_\_ JOB NO. \_\_\_\_\_

PURPOSE \_\_\_\_\_ SHEET 2 OF 4

COMPUTED BY \_\_\_\_\_ CHECKED BY \_\_\_\_\_ DATE \_\_\_\_\_

Example Opening Span 32'; 91' gravel overburden; 10' clay  $\gamma = 105 \text{ lb/ft}^3$ a) Shale beam: 18" thick  $E = 1 \times 10^6 \text{ psi}$  tensile strength 300 psi  
 $\gamma = 140 \text{ lb/ft}^3$ b) sandstone beam: 7" thick  $E = 5 \times 10^6 \text{ psi}$  tensile strength 500 psi  
 $\gamma = 150 \text{ lb/ft}^3$  w/ 91' gravel overburden + 10' clay overburden

c) Assume no tensile strength between bedding calc deflections, is self-loading.

Shale beam

$$q = 140 \text{ lb/ft}^3 (18") \cdot 1' = 2520 \text{ lbs/ft length}$$

$$I = \frac{bh^3}{12} = \frac{1(18)^3}{12} = 486 \quad \text{Max defl. } x = L/2$$

$$\text{deflection } W = \frac{q \cdot x^2 (1-x)^2}{24EI} \quad ; \quad W = \frac{q \cdot L^4 (1/2)^2}{24EI}$$

$$W = \frac{2520 \left( \frac{32^4}{4} \right) \left( \frac{1}{2} \right)^2}{24 (1 \times 10^6 \text{ psi} \cdot \frac{144 \text{ in}^4}{12}) (486)} = 2.1966 \times 10^{-5} \text{ ft}$$

Sandstone beam w/ weight of clay and gravel overburden

$$q = 91' \cdot 140 \text{ lb/ft}^3 + 10' \cdot 105 \text{ lb/ft}^3 + 7' \cdot 150 \text{ lb/ft}^3 \cdot 1 = 15295 \text{ lbs/ft beam}$$

$$I = \frac{1 \left( \frac{7}{12} \right)^3}{12} = 28.583 \quad \text{defl. at } x = L/2$$

$$W = \frac{q \cdot L^4 (1/2)^2}{24EI}$$

$$W = \frac{15295 \text{ lbs} \left( \frac{32^4}{4} \right) \left( \frac{1}{2} \right)^2}{24 (5 \times 10^6 \text{ psi} \cdot \frac{144 \text{ in}^4}{12}) (28.583)} = 4.5 \times 10^{-4} \text{ ft}$$

Sandstone beams deflects more than shale beams so beams are self loading to act as composite

For this condition if there is no slippage between bedding planes can analyze problem as simple composite beam

\* If sliding occurs between bedding planes and sandstone and shale then:

$$\frac{M}{EI_{\text{shale}}} = \frac{q(L^2 - 6Lx + 6x^2)}{[E_s I_s + E_{\text{sh}} I_{\text{sh}}]} \quad \text{with Max tension at } x = 0 \text{ L=0}$$

where  $\frac{1}{EI}$  of layers = deflections

Will SHALE fail in tension?





PROJECT \_\_\_\_\_ JOB NO. \_\_\_\_\_

PURPOSE \_\_\_\_\_ SHEET 3 OF 4

COMPUTED BY \_\_\_\_\_ CHECKED BY \_\_\_\_\_ DATE \_\_\_\_\_

$$M = \frac{(1.44 \times 10^6)(486)}{17815 \text{ lb} \left( \frac{22^2}{2} - 0 + 0 \right)} = \frac{17815 \text{ lb} (22^2 - 0 + 0)}{[1.44 \times 10^6 (16)^2 + 7.2 \times 10^6 (7)^2]} \Rightarrow \frac{8622460}{1.086768 \times 10^{12}}$$

$$q = 71' \cdot 145 + 10' \cdot 105 + 7' \cdot 130 + 18' \cdot 145 = 17815 \text{ lbs / length beam}$$

$$\frac{M}{I} = \frac{7.93403728 \times 10^{-12}}{6.975 \times 10^{-10}} \quad M = 555256 \quad 10^{-11}$$

$$\sigma_{\text{max}} = \frac{Mc}{I} \quad \sigma_{\text{max}} = \frac{555256 (18/2)}{486} = 10282.5 \frac{\text{lb}}{\text{in}^2} = 71 \text{ psi}$$

$$FS = \frac{300 \text{ psi lab tensile strength}}{71 \text{ psi beam}} = 4.2$$

If case of shale beam self loading only, that is shale deflects more than sandstone above producing bed separation between the two; then Tensile failure analysis is

$$\sigma_{\text{max tensile}} = \frac{qL^2}{2bh^2}$$

$$q = 140 \text{ lb/ft}^2 \cdot 18 \text{ ft} = 2520 \text{ lbs / ft length}$$

$$\sigma_{\text{max}} = \frac{2520 (18)^2}{2 \cdot 1 \cdot 18^2} = 1862 \text{ lb/in}^2 \quad \text{or} \quad 13.1 \text{ psi}$$

$$FS = \frac{300 \text{ psi}}{13.1} = 23$$



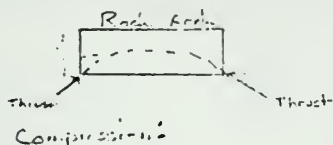
PROJECT Red Lodge / Beauregard

JOB NO. \_\_\_\_\_

PURPOSE Analyze According to Cox w/ Vauissier Arch  
for compressional and shear failureSHEET 4 OF 4

COMPUTED BY \_\_\_\_\_ CHECKED BY \_\_\_\_\_

DATE \_\_\_\_\_



1) Consider shale beam 18 ft. thick  
with 125' of rock & overburden, span 22'  
 $q_u = 435$  psi assumed No cohesion along bedding =  $S_0$   
 $\phi = 20^\circ$

where from above  $\gamma_h = 17815$  lbs/ft<sup>2</sup>  
 $t$  = Beam thickness

$$H = \frac{17815(22)^2}{3 \frac{3}{4} (18)} = 79337.6 \text{ lbs/ft} = \text{554 kips}$$

$$C = \frac{H}{t/4}$$

$$C = \frac{79337.6}{18/4} = 17741.7 \text{ lbs/ft}^2 \text{ or } 123 \text{ psi}$$

C = compressive  
strength beam

$$FS = \frac{435}{123} = 3.5 \rightarrow \text{For Compressional failure}$$

Calc min beam thickness beam required for stability

$$t = \sqrt{\frac{\gamma_h L}{3/2 C}}$$

$$t = \sqrt{\frac{17815 \text{ ft} (22)}{3/2 (435 \text{ psi}) (144 \text{ psi/ft}^2)}}$$

$$t = 9.6'$$

where  
C = compressive  
strength beam

$$FS = \frac{13}{9.6} = 1.9$$

Shear Failure:

$$V = \frac{\gamma_h L}{t/4}$$

$$V = \frac{17815(22)}{18/4} = 43547.8 \text{ lbs/ft} = 3.5 \text{ kips}$$

$$\frac{1/2 \times 12 \times 12}{2 \times 12} = \frac{1}{2}$$

$$S = S_0 + \sigma' \tan \phi$$

Same as (2) =  $C + \sigma' \tan \phi$

$$S = 0 + 17741.7 \tan 20^\circ = 10242.2 \text{ psi mobilized}$$

$$FS = \frac{10242.2}{43547.8} = 0.23$$

which indicates arching forces don't always  
prevent minor block falls in roof

$$FS = \frac{\text{1 arching force}}{\text{Driving force}}$$





RED LIDGE / BEHARDRECK SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: No.1 1/2 Seam at D-1  
Room Size, Ft. = 22

Layer No.	Unit Abb.	Layer Thickness Ft.	E psi	Tensile Strength psi	Individual Layer Deflection Ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure	BUY(D203)
1	145	5.			0.20E+22	0.2	=DIV/0	1.32E+24
2	125	12			=DIV/0	2.2	=DIV/0	1.42E+24
3	132	7	4E+25	352	3.87E+23	11.8	0.8	1.53E+24
4	142	15	4E+25	352	3.75E+23	32.4	4.9	1.76E+24



RED LIPS / BECDRECK EMBROIDERY STUDY  
 Roof Collapse Analysis

Roof Description: No. 1 1/2 Seam at D4-1  
 Load Case, F.L. = 15

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure	SUM TOTAL
1	145	5.			0.00E+00	0.0	#DIV/0!	1.32E+04
2	125	12			#DIV/0!	0.0	#DIV/0!	1.42E+04
3	150	7	4E+05	350	9.45E-03	7.5	2.5	1.53E+04
4	140	13	4E+05	350	9.15E-03	20.7	3.1	1.78E+04



RED CROSS / BEARINGS SURVEILLANCE STUDY  
Roof Collapse Analysis

Problem Description: No. 1 1/2 Seam at Dn-1  
Room Size, Ft. = 32

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure	SUM(CR203)
1	145	5			0.00E+00	2.0	#DIV/0!	1.35E+24
2	125	2			#DIV/0!	2.0	#DIV/0!	1.45E+24
3	150	7	4E+05	350	1.96E-23	5.2	2.4	1.53E+24
4	140	10	4E+05	350	1.90E-24	14.4	2.2	1.73E+24



RED LODE / BEARDREE SUBSIDENCE STUDY  
 Roof Collapse Analysis

Problem Description: No.1 1/2 Seam at DM-1  
 Seam Size, ft. = 35

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S.	F.S.	SUM(MODS)
						Individual Layer Tensile Failure	Composite Layer Tensile Failure	
1	145	91			0.22E+22	0.0	#DIV/0	1.32E+24
2	125	10			#DIV/0!	2.2	#DIV/0	1.42E+24
3	133	7	4E+25	352	2.63E+22	3.8	2.3	1.53E+24
4	142	13	4E+25	350	3.52E+24	10.6	1.5	1.78E+24





RED LODGE / BEAR CREEK SUBSIDENCE STUDY  
Roof Collapse Analysis

Roof Description: No. 1 1/2 Seam at D-1  
Room Size, ft. x ft. = 30

Layer No.	Unit Abbrev.	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure	SUM(CDSD3)
1	143	91			0.20E+00	0.0	#DIV/0	1.33E+04
	125	10			#DIV/0	0.0	#DIV/0	1.42E+04
	152	7	4E+05	175	3.67E-03	5.9	0.4	1.53E+04
4	140	18	4E+05	175	3.75E-03	16.2	2.4	1.72E+04



RED LODE / SEAROCK CREEK SUBSIDENCE STUDY  
Rock Collapse Analysis

Problem Description: No. 1 1/2 Seam at D#-1  
Room Size, Ft. = 25

Layer No.	Unit Weight pcf	Layer Thickness Ft.	E psi	Tensile Strength psi	Individual Layer Deflection Ft.	F.S.	F.S.	SUM(C02C3)
						Individual Layer Tensile Failure	Composite Layer Tensile Failure	
1	145	31			0.00E+00	0.0	#DIV/0!	1.38E+04
2	125	12			#DIV/0!	0.0	#DIV/0!	1.48E+04
3	150	7	4E+05	175	9.45E-03	3.8	0.3	1.53E+04
4	142	19	4E+05	175	9.18E-03	10.4	1.6	1.73E+04



150 L003 / BEARPEAK SURVIVENCE STUDY  
Roof Collapse Analysis

Problem Description: No.1 1/2 Seam at B-1  
Seam Size, ft.= 30

Layer No.	Unit Weight	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection	F.E. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure	SUM(C203)
					ft.			
1	145	9.			0.00E+00	0.2	#DIV/0!	1.32E+24
2	125	10			#DIV/0!	2.0	#DIV/0!	1.42E+24
3	150	7	4E+25	175	1.96E+20	2.6	0.2	1.53E+24
4	140	18	4E+25	175	1.50E+24	7.2	1.1	1.75E+24



RED LIME / REARDEEN SUBSIDENCE STUDY  
Roof Collapse Analysis

Roof Description: No.1 1/2 Seam at B--1  
Room Size: ft. = 15

Layer No.	Unit Weight	Layer Thickness	E	Tensile Strength	Individual Layer	F.S. Individual Layer	F.S. Composite Layer	SUM(CROSS)
					Deflection	Tensile Failure	Tensile Failure	
	pcf	ft.	psi	psi	ft.			
1	145	5.			0.00E+00	0.0	#DIV/0!	1.32E+24
2	125	10			#DIV/0!	0.0	#DIV/0!	1.42E+24
3	155	7	4E+05	175	3.63E-23	1.9	0.1	1.53E+24
4	140	13	4E+05	175	3.62E-24	5.0	0.6	1.73E+24





RED LODGE / BEARCREAK SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 1 1/2 Seam at 5m-1  
Room Size, ft. = 10

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S.	F.S.	SUM(C203)
						Individual Layer Tensile Failure	Composite Layer Tensile Failure	
1	145	5.1			0.00E+00	0.0	#DIV/0!	1.30E+04
2	135	1.2			#DIV/0!	0.0	#DIV/0!	1.40E+04
3	150	7	4E+08	88	3.67E-05	3.0	0.2	1.50E+04
4	140	18	4E+08	88	3.75E-05	8.1	1.2	1.70E+04



RED LODGE / BERNARD CREEK SURVEILLANCE STUDY  
 Roof Collapse Analysis

Problem Description: No.1 1/2 Seam at D--1  
 Room Size: 20' x 22'

Layer No.	Unit Weight pcf	Layer Thickness in.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S.	F.S.	S.M. #106803
						Individual Layer Tensile Failure	Composite Layer Tensile Failure	
1	145	51			0.00E+00	0.0	#DIV/0	1.32E+04
2	125	12			#DIV/0	0.0	#DIV/0	1.42E+04
3	150	7	4E+08	88	9.42E-03	1.0	0.1	1.53E+04
4	140	19	4E+08	88	9.15E-03	5.1	2.5	1.73E+04



# RED LINE / SEAWALL STABILITY STUDY Red Collapse Analysis

Problem Description: No. 1 1/2 Seam at D--1  
 Seam Size, ft. = 30

Layer No.	Unit No. 100	Layer Thickness ft.	E ksi	Tensile Strength ksi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure	SDV(DSDI)
1	145	31			0.00E+00	0.0	#DIV/0!	1.38E+24
2	125	2			#DIV/0!	0.0	#DIV/0!	1.43E+24
3	150	7	4E+05	88	1.95E-22	1.2	0.1	1.53E+24
4	140	10	4E+05	88	1.90E-24	3.5	0.5	1.75E+24



RED LODGE / BEARCRICK SUBSIDENCE STUDY  
Roof Delineate Analysis

Problem Description: No. 1 1/2 Seam at D-1  
Room Size, ft. = 32

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure	SUM(DEFCT)
145		31			0.00E+00	0.0	#DIV/0!	1.32E+04
135		10			#DIV/0!	0.0	#DIV/0!	1.42E+04
130		7	22-25	350	7.74E-03	11.8	0.8	1.53E+04
120		10	22-25	350	7.50E-05	32.4	4.9	1.78E+04





FE0 10001 / SEAFORCE4 GUIDANCE STUDY  
 Top Collapse Analysis

Problem Description: No.1 1/2 Seam at D-1  
 Seam Size, ft. = 30

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure	SUM(CROSS)
1	145	5.1			2.20E-23	2.2	#DIV/0	1.32E+04
2	125	1.2			#DIV/0	0.2	#DIV/0	1.42E+04
3	150	7	1E+05	350	1.55E-22	11.8	0.6	1.53E+04
4	140	.8	1E+05	350	1.52E-24	33.4	4.5	1.75E+04



RED LODGE / BEAVER CREEK SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: Seam No. 2 at D#-1  
Room Size, ft. = 20

Layer No.	In. of Roof	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection in.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	14.1	.21	2E+22	0	2.22E-22	2.0	#DIV/0!
2	53	.25	2E+22	0	#DIV/0!	2.0	#DIV/0!
3	53	.25	4E+25	250	1.74E-25	227.5	118.0
4	53	.25	4E+25	300	7.20E-25	32.4	1216.2



RED LODGE / BEARCREEK SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: Seam No. 2 at D-1  
Room Size, ft. = 25

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.E. Individual Layer Tensile Failure	F.E. Composite Layer Tensile Failure
1	141	121	8E+22	0	0.00E+00	2.2	#DIV/0!
2	83	35	0E+00	0	#DIV/0!	0.3	#DIV/0!
3	83	106	4E+05	350	4.25E-05	225.9	74.4
4	83	12	4E+05	350	1.75E-24	19.4	775.7



RED LODGE / BEARCREAK SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: Seam No. 2 at D#1  
Room Size, ft. = 30

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	141	101	2E+22	2	2.22E-22	2.2	#DIV/0!
2	83	25	2E+22	2	#DIV/0!	2.2	#DIV/0!
3	83	105	4E+25	352	6.62E-25	145.7	51.7
4	83	12	4E+25	352	3.65E-24	12.5	542.3





RED LODGE / BEAVER CREEK SUBSIDENCE STUDY  
 ROC<sup>2</sup> Collapse Analysis

Problem Description: Seam No. 2 at D-1  
 Panel Size, ft. = 32

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	141	101	0E+00	0	0.00E+00	0.0	#DIV/0
2	83	25	0E+00	0	#DIV/0!	0.0	#DIV/0!
3	83	106	4E+05	350	1.74E-25	327.9	116.3
4	81	5	4E+05	350	2.89E-24	15.2	2471.1



RED LODGE / BEARCREEK SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: Seam No. 2 at D-1  
Room Size, ft. = 20

Layer No.	Unit weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	141	121	2E+22	0	0.22E-22	0.0	#DIV/0
2	63	25	2E+22	0	#Div/0	0.0	#Div/0
3	63	108	4E+26	252	1.74E-26	327.5	116.2
4	63	2	4E+26	352	1.82E-23	5.1	6237.7



RED LODGE / BEAVERCREEK SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: Seam No. 2 at D4-1  
Room Size, ft. = 25

Layer No.	Unit Weight ccf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	141	101	2E+22	0	0.00E+00	0.0	#DIV/0!
2	85	25	0E+00	0	#DIV/0!	0.0	#DIV/0!
3	85	108	4E+25	352	4.25E-25	203.5	74.4
4	85	1	4E+25	352	1.75E-25	1.9	8212.4



RED LODGE / BEAVERHEAD RESIDENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 4 Seam at D-3  
Room Size: 7' x 22'

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer	F.S. Individual Layer	F.S. Composite Layer
					Deflection ft.	Tensile Failure	Tensile Failure
1	85	180	8E+03	2	0.00E+00	0.0	#DIV/0!
2	97	104	4E+05	350	1.85E-05	321.2	113.5
3	92	14	7E+05	650	2.55E-05	76.2	894.1
4					#DIV/0!	#DIV/0!	#DIV/0!





RED LOOSE / BERRICKREEK SUBSIDENCE STUDY  
 Prof Collapse Analysis

Model Description: No. 4 Seam at B-3  
 Seam Size, ft. = 35

Layer No.	Unit Weightpcf	Layer Thicknessft.	Epsi	Tensile Strengthpsi	Individual Layer Deflectionft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	83	1.50	25*20	3	2.08E-02	0.0	#DIV/0!
2	87	1.04	45*25	350	4.92E-03	193.2	73.7
3	92	1.4	75*25	550	5.56E-03	48.7	575.3
4					#DIV/0!	#DIV/0!	#DIV/0!



SEE LOGS / BEARDREEK SUBSIDENCE STUDY  
 Rod Collapse Analysis

Problem Description: No. 4 Seam at D-3  
 Rod Size, ft.: 32

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	55	182	2E+02	2	0.00E+00	0.0	#DIV/0!
2	87	104	4E+05	350	9.37E-25	133.5	521.5
3	90	14	7E+05	680	1.15E-24	33.5	357.4
4					#DIV/0!	#DIV/0!	#DIV/0!



RED LODGE / BEARCREAK RESURGENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 4 Seam at D-3  
Seam Size, ft. = 43

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	83	1.2	8E+22	0	0.00E+00	0.0	#DIV/0!
2	97	1.04	4E+23	350	2.95E-25	75.2	23.4
3	80	14	7E+25	650	3.64E-24	19.0	223.5
					#DIV/0!	#DIV/0!	#DIV/0!



RED LODGE / BEARCREEK SUBSIDENCE STUDY  
Roof Collapse Analysis

Roof Description: No. 4 Seam at D-3  
Box Size, ft. = 32

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S.	F.S.
						Individual Layer Tensile Failure	Composite Layer Tensile Failure
1	85	1.02	0E+00	0	0.00E+00	0.0	#DIV/0
2	87	1.04	4E+24	350	1.85E-25	381.2	113.5
3	92	1.4	7E+24	680	2.28E-24	76.2	694.1
4					#DIV/0	#DIV/0	#DIV/0





FEED 10018 / SEAWATER SUBSIDENCE STUDY  
 Spill Collage Analysis

Problem Description: No. 4 Seam at 2-3  
 Seam Size, ft. 20

Layer No.	Unit Weightpcf	Layer Thicknessft.	Epsi	Tensile Strengthpsi	Individual Layer Deflectionft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	83	158	2E+22	3	2.00E+22	2.2	#DIV/0!
2	87	124	4E+24	175	1.85E+25	150.5	55.5
3	90	14	7E+24	680	2.23E+24	76.2	854.1
4					#DIV/0!	#DIV/0!	#DIV/0!



450 L0088 / BEARINGS SUBSISTENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 4 Seam at D--2  
Room Size, ft. = 20

Layer No.	Unit Weightpcf	Layer Thicknessft.	Epsi	Tensile Strengthpsi	Individual Layer Deflectionft.	F.S.	F.S.
						Individual Layer Tensile Failure	Composite Layer Tensile Failure
1	83	180	0E+00	0	0.00E+00	0.0	#DIV/0
2	87	104	4E+04	350	1.85E-05	301.5	113.5
3	82	14	7E+04	340	2.88E-04	35.1	447.1
4					#DIV/0	#DIV/0	#DIV/0



RED LODGE / BEARCREAK SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 4 Seam at D#-3  
Room Size, ft. = 30

Layer No.	Unit Weightpcf	Layer Thickness Ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	83	183	85E+00	0	2.20E-22	0.2	#DIV/0
2	97	24	4E+04	175	1.85E-25	150.6	58.8
3	50	14	7E+24	342	2.86E-24	38.1	447.1
4					#DIV/0	#DIV/0	#DIV/0



RED LODGE / BEARCREEK GLEISSENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 4 Seam at DH-3  
Room Size, Ft. = 38

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.E. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	65	1.83	2E+00	0	0.00E+00	0.0	#Div/20
2	87	1.0	4E+04	175	9.37E-05	65.9	25.1
3	90	3	7E+04	342	2.51E-02	3.8	55.0
4					#Div/0	#Div/20	#Div/20





RED LOOSE BEACHHEAD SURSISTENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 4 Seam at D--3  
Room Size, ft.<sup>2</sup> 32

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	81	162	2E+22	0	0.00E+00	0.0	#DIV/0
2	87	124	4E+24	175	9.37E-25	65.9	25.3
3	92	2	7E+24	240	5.55E-22	2.4	1445.9
4					#DIV/0	#DIV/0	#DIV/0



RED LODES DEBRIDGE SUBSIDENCE STUDY  
 Rock Collapse Analysis

Problem Description: No. 4 Seam at Dn-3  
 Seam Size, ft. = 30

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	53	162	8E+08	0	3.03E-08	2.0	#Div/0
2	27	124	4E+04	175	9.37E-05	66.9	35.1
3	92	1	7E+04	348	2.25E-01	1.2	2904.5
4					#Div/0	#Div/0	#Div/0



FE0 0000 / SEAROSEK SLENDENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 4 Seam at D-3  
Room Size, ft. = 32

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	63	192	8E+20	0	0.02E+20	0.0	#DIV/0!
2	57	124	4E+24	350	9.37E+25	133.9	50.5
3	52	1	7E+24	450	2.26E+21	2.4	5829.2
4					#DIV/0!	#DIV/0!	#DIV/0!



RED LODGE / BEPPOWEEK SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 5 Seam at 24-3  
Room Size, ft. = 20

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	54	333	25-30	0	3.02E-02	3.3	=DIV/0
2	83	152	45-25	350	4.53E-02	552.6	242.7
3	93	12	45-25	270	6.07E-05	20.9	3331.1
4	93	15	35-25	150	7.17E-05	17.4	1.1





## RED LODGE REFORMER SUBSIDENCE STUDY

## Roof Collapse Analysis

Problem Description: No. 5 Seam at DM-3  
 Section Size, ft. = 21

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
						Failure	Failure
1	84	3.55	33E+24	0	0.00E+00	0.0	#DIV/21
2	83	1.33	4E+25	350	1.33E-25	353.7	155.3
3	83	1.0	4E+25	270	1.57E-24	13.4	2131.5
4	81	15	2E+25	150	1.75E-24	11.1	0.7



400 L000 / BEARCRICK RESIDENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 5 Seam at J4-3  
Room Size, ft. = 50

Layer No.	Layer Weight lb/sq ft.	Layer Thickness in.	Tensile Strength psi	Individual Layer Deflection in.	F.E. Individual Layer Tensile Failure	F.E. Composite Layer Tensile Failure
1	64	30	35-23	2	2.22E-23	2.2
2	83	18	45-28	333	1.52E-23	1.5
3	53	12	45-28	370	4.25E-24	4.3
4	53	15	55-28	150	3.63E-24	3.7



FED LOOSE / BEAR CREEK SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 3 Seam at DM-3  
Room Size, Ft. = 35

Layer No.	Unit Weight: pcf	Layer Thickness: ft.	E psi	Tensile Strength: psi	Individual Layer Deflection: ft.	F.S.	
						Individual Layer Tensile Failure	Composite Layer Tensile Failure
1	84	223	2E+00	2	0.00E+00	2.0	400/0
2	80	188	4E+05	350	4.50E-05	100.4	75.1
3	80	10	4E+05	370	7.57E-04	5.8	1007.7
4					#DIV/0!	#DIV/0!	#DIV/0!



RED 1033 / BARRAGE SUBSIDENCE STUDY  
Roof Collapse Analysis

Roof Description: No. 5 seam at D--3  
Seam Size, ft. = 20

Layer No.	Unit Weight pcf	Layer Thickness ft.	Tensile Strength psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S.	F.S.
						Individual Layer Tensile Failure	Composite Layer Tensile Failure
1	145	2.0	2E+02	2	0.20E-02	2.0	4077/0
2	145	1.0	4E+02	100	5.39E-07	315.3	139.5
3	155	.2	4E+02	270	1.34E-04	12.5	1314/-
4	155	.5	2E+02	150	1.22E-04	12.5	0.8





RED LODGE / BEAVERCREEK SUBSIDENCE STUDY  
 Roof Collapse Analysis

Problem Description: No. 5 seam at D4-3  
 Room Size, ft. = 30

Layer No.	Unit Height ft.	Layer Thickness ft.	E ksi	Tensile Strength ksi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	1.45	235	25-20	2	0.02E-26	0.0	#DIV/0
2	1.45	182	42-25	351	6.59E-27	315.3	135.5
3	1.55	12	42-25	270	1.34E-24	12.5	194.4
4	1.55	15	25-25	250	1.20E-24	17.4	1.0



RED WOODS / BEARWOOD SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 5 seam at Dn-3  
Room Size, Ft. = 35

Layer No.	Unit weight, $\text{pcf}$	Layer Thickness, Ft.	E, psi	Tensile Strength, psi	Individual Layer Deflection, Ft.	F.S.	F.S.
						Individual Layer Tensile Failure	Composite Layer Tensile Failure
1	146	269	$2E+20$	0	$2.00E+02$	3.0	#D1//2
2	146	132	$4E+25$	330	$8.26E-25$	123.2	45.7
3	155	10	$4E+25$	270	$1.26E-23$	4.1	625.1
4	155	15	$3E+25$	150	$1.12E-23$	3.4	2.2



RED 1000 / GERRARD RESIDENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 5 Seam at C-3  
Roof Size, ft. = 35

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection in.	F.E. Individual Layer Tensile Failure	F.E. Composite Layer Tensile Failure
84	200	2E+02	2	2.00E+00	2.0	#DIV/0!	
83	180	4E+04	350	4.00E+03	180.0	75.0	
82	90	1E	4E+04	270	7.57E+00	0.0	1007.7
4					#DIV/0!	#DIV/0!	#DIV/0!



RED LODGE / BEAVERHEAD SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 5 Seam at D-1  
Room Size, ft. = 32

Layer No.	Unit Weight pcf	Layer Thickness ft.	=	Tensile Strength psi	Individual Layer Reflection %	F.E. Individual Layer Tensile Failure	F.E. Composite Layer Tensile Failure
1	84	0.05	=	68+20	2	0.005+0.0	0.0
2	160	.02	=	45+20	175	4.000+25	36.2
3	80	.10	=	45+20	175	7.575+25	3.4
4						0.014/2	0.014/2





RED LODE / BEHRENDEN SUBSIDENCE STUDY  
100' Collar Analysis

Problem Description: No. 3 Seam at Dm-3  
Coal Size, ft. = 15

Layer No.	Unit Weight pcf	Layer Thickness ft.	E psi	Tensile Strength psi	Individual Layer Deflection ft.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	64	2.00	25-20	2	0.005-00	3.2	#016/2
2	80	1.00	#01-20	100	1.825-00	333.7	133.3
3	93	"	#01-20	270	7.855-00	8.7	433.3
4					#016/2	#017/2	#017/2



RED LODGE / BEAR CREEK SUBSIDENCE STUDY  
Roof Collapse Analysis

Problem Description: No. 3 Seam at D4-3  
Room Size: 17' x 30'

Layer No.	Unit Weight pcf	Layer Thickness ft.	Tensile Strength psi	Individual Layer Deflection in.	F.S. Individual Layer Tensile Failure	F.S. Composite Layer Tensile Failure
1	64	0.05	0	2.22E-07	3.0	#Div/0
2	93	0.02	370	1.32E-05	333.7	333.7
3	93	0.02	270	4.98E-05	2.7	10865.2
4				#Div/0	#Div/0	#Div/0









# Northern

Engineering  
and Testing, Inc.

## STANDARD COMPUTATION SHEET

PROJECT Red Lodge / Bonanza JOB NO. 87-3001-D-5  
PURPOSE Coles 1 1/2 seam DH-1; Pillar loads SHEET 1 OF 2  
COMPUTED BY FD CHECKED BY J. Paul DATE 8-21-87

Pillar #1

$$P = \frac{(L_p + W_o)(W_p + W_o) \gamma_h}{W_p L_p}$$

Seam Ht = 5.5 ft

$L_p = 55'$  } Pillar Dimensions

$W_p = 25'$

$W_o + L_p = 80'$  length of trib. area

$\gamma_{seam} = 145 \text{ lb/ft}^3 \cdot 9'$

$\gamma_{seam} = 150 \text{ lb/ft}^3 \cdot 36'$

$W_o + W_p = 42'$  width of Trib. Area

$$P = \frac{(80)(42) \cdot 18595}{25 \cdot 55} = 45440 \frac{\text{lbs}}{\text{ft}^2} \text{ or } 315.6 \text{ psi}$$

Pillar #2

$L_p = 55'$

$W_p = 32'$

$L_p + W_o = 80'$

$W_p + W_o = 55'$

$\gamma_h = 18595$

$$P = \frac{(80)(55) \cdot 18595}{55 \cdot 32} = 49393 \frac{\text{lbs}}{\text{ft}^2} \text{ or } 343 \text{ psi}$$

Pillar #3

$$P = \frac{(80)(50)(18595)}{50 \cdot 25} = 42355 \frac{\text{lbs}}{\text{ft}^2} \text{ or } 294.1 \text{ psi}$$

$L_p = 70'$

$W_p = 25'$

$L_p + W_o = 82'$

$W_p + W_o = 50'$

$\gamma_h = 18595$

Pillar #4

$$P = \frac{(125)(55)(18595)}{115 \cdot 30} = 37055 \frac{\text{lbs}}{\text{ft}^2} \text{ or } 257 \text{ psi}$$

$L_p = 115'$

$W_p = 30'$

$L_p + W_o = 125'$

$W_p + W_o = 55'$

$\gamma_h = 18595$

Pillar #5

$$P = \frac{(68)(68)(18595)}{57 \cdot 31} = 46513 \frac{\text{lbs}}{\text{ft}^2} \text{ or } 323 \text{ psi}$$

$L_p = 57'$

$W_p = 31'$

$L_p + W_o = 63'$

$W_p + W_o = 65'$

$\gamma_h = 18595$







## STANDARD COMPUTATION SHEET

PROJECT Red Lodge / Benchmark JOB NO. 87-3001-D-5  
 PURPOSE Calcs 1 1/2 sec. D44 SHEET 2 OF       
 COMPUTED BY      CHECKED BY      DATE 8-21-57

Pillar #6  
 $L_p = 50'$   
 $W_p = 25'$   
 $L_w + W_h = 90'$   
 $W_p + W_h = 50'$   
 $A_H = 1800'$

$$P = \frac{(90)(50) 1800'}{(25)(85)} = 39378 \text{ lbs/ft}^2 \text{ or } 273.5 \text{ psi}$$

Pillar #7  
 $W_p = 45'$   
 $L_w + W_p = 50'$   
 $A_H = 1800'$

$$P = \frac{(W_h + W_p) A_H}{W_p^2}$$

$$P = \frac{(50') 1800'}{(45)^2} = 25774 \text{ lbs/ft}^2 \text{ } 179.1 \text{ psi}$$

Strength of conc. cubes Aves. 2500 psi

Correction for size of cubes:

$$K = \sigma_c \sqrt{D} = (2500 \text{ psi}) \sqrt{2}$$

$$K = 3535$$

correction size of pillars:  $\sigma_1 = \frac{K}{C} = \frac{3535}{6} \approx 590 \text{ psi}$

Correction for pillar dimensions =  $0.64 + 0.36 \frac{W}{H}$

where  $W/H$  = pillar width to height ratio

Pillar No.	Applied Stress, psi	Pillar $W/H$	Strength of Pillar, psi	Soil Factor
1	316	$\frac{25}{50} = 4.6$	1355	4.3
2	343	5.8	1541	4.5
3	294	4.6	1300	4.4
4	257	5.5	1546	6.0
5	323	5.6	1567	4.9
6	277	4.6	1355	4.9
7	179	8.2	2119	11.8



RED LODGE / BEARCREEK SLURRIENCE STUDY  
Column Crushing Analysis

Problem Description: No.1 1/2 Seam at D-1

Coal Gu. (cst) = 2500 Seam Thickness, ft. = 5  
Pillar Dia. or Size, in. = 2

PILLAR No.	PILLAR DIMENSIONS		TRIBUTARY AREA		OVERBURDEN	PILLAR STRESS	SAFETY FACTOR	PILLAR W/F RATIO	PILLAR STRENGTH
	Length ft.	Width ft.	Length ft.	Width ft.	PSF	PSI			PSI
1	55	25	55	45	18555	315	3.3	2.1	1242
2	55	25	55	55	18555	342	3.5	4.2	1225
3	75	35	55	55	18555	254	3.5	3.1	1242
4	115	35	125	55	18555	257	4.6	3.8	1173
5	55	21	55	55	18555	323	3.7	2.9	1199
6	65	25	55	55	18555	321	3.5	3.1	1242
7	45	45	55	55	18555	179	8.5	5.5	1572
8					18555	#DIV/0!	#DIV/0!	0.0	377
9					18555	#DIV/0!	#DIV/0!	0.0	377
10					18555	#DIV/0!	#DIV/0!	0.0	377
11					18555	#DIV/0!	#DIV/0!	0.0	377



RED LODGE / BEAUFORT CESSIDENCE STUDY  
Column Crushing Analysis

Problem Description: No.1 1/2 Seam at D4-1

cal. Su. (psi) = 2228 Seam Thickness, ft. = 8  
u Dia. or Size, in. = 2

PILLAR No.	PILLAR DIMENSIONS		TRIBUTARY AREA		OVERLAP STRESS psi	PILLAR STRESS psi	SAFETY FACTOR	PILLAR W/F RATIO	PILLAR STRESS psi
	Length ft.	Width ft.	Length ft.	Width ft.					
1	55	25	80	42	18595	216	2.6	2.1	612
2	55	30	85	50	18595	343	2.9	4.2	551
3	70	35	85	50	18595	394	2.8	3.1	608
4	110	30	125	55	18595	257	3.5	1.8	319
5	57	31	85	60	18595	323	2.2	3.5	555
6	65	35	90	55	18595	321	2.8	3.1	608
7	45	45	55	53	18595	173	7.0	5.5	1258
8					18595	#DIV/0!	#DIV/0!	0.0	102
9					18595	#DIV/0!	#DIV/0!	0.2	322
10					18595	#DIV/0!	#DIV/0!	0.0	322
11					18595	#DIV/0!	#DIV/0!	0.2	322



RED LODGE / BEAR CREEK SLURRY STUDY  
Column Crushing Analysis

Problem Description: No. 1 1/2 Seam at Dm.

Coal Du. (psi) = 1500 Seam Thickness, ft. = 5  
Du Dia. or Size, in. = 2

PILLAR	PILLAR DIMENSIONS		TWO BUTTARY OVER		OVERLAP OVER		SAFETY FACTOR	PILLAR	PILLAR
No.	Length ft.	Width ft.	Length ft.	Width ft.	STRESS psi	STRESS psi		W/ RATIO	STRESS psi
1	55	55	55	42	16555	315	2.0	3.0	634
2	55	55	55	55	16555	343	2.0	4.0	732
3	70	55	55	55	16555	294	2.0	3.0	634
4	115	55	155	55	16555	257	2.0	3.0	704
5	155	55	155	55	16555	322	2.0	3.0	715
6	155	55	155	55	16555	321	2.0	3.0	634
7	155	55	155	55	16555	175	5.0	5.0	942
8	155	55	155	55	16555	#DIV/0!	#DIV/0!	2.0	235
9	155	55	155	55	16555	#DIV/0!	#DIV/0!	2.0	235
10	155	55	155	55	16555	#DIV/0!	#DIV/0!	2.0	235
11	155	55	155	55	16555	#DIV/0!	#DIV/0!	2.0	235





RED LODGE / BEAVERCREEK SUBSIDENCE STUDY  
Column Crushing Analysis

Problem Description: No. 1 1/2 Seam at D4-1

Coal Gv. (psi) = 1222 Seam Thickness, ft. = 3  
Pillar Dia. or Size, in. = 2

PILLAR No.	PILLAR DIMENSIONS		TRIANGULAR AREA		OVERBURDEN	PILLAR STRESS	SAFETY FACTOR	PILLAR W/H RATIO	PILLAR STRENGTH
	Length ft.	Width ft.	Length ft.	Width ft.	psf	psi			psi
1	55	25	60	42	18595	315	1.3	3.1	416
2	55	25	60	33	18595	343	1.4	4.2	492
3	72	25	55	50	18595	294	1.4	3.1	416
4	115	20	125	55	18595	257	1.8	3.8	483
5	57	21	66	67	18595	320	1.5	3.5	482
6	65	25	90	55	18595	321	1.4	3.1	416
7	48	25	22	55	18595	179	3.5	5.6	688
8					18595	#DIV/0!	#DIV/0!	0.2	151
9					18595	#DIV/0!	#DIV/0!	0.2	151
10					18595	#DIV/0!	#DIV/0!	0.2	151
11					18595	#DIV/0!	#DIV/0!	0.2	151



RED LODGE / BEAR CREEK SUBSIDENCE STUDY  
Column Crushing Analysis

Problems Description: No. 2 Seam at D-1

Coal Du. (psi) = 8522 Bean Thickness, ft. = 8  
Du Dia. or Size, in. = 2

Pillar No.	PILLAR DIMENSIONS		TRIANGULAR AREA		OVERLOADING	PILLAR STRESS	SAFETY FACTOR	PILLAR W/10	PILLAR STRESS
	Length ft.	Width ft.	Length ft.	Width ft.	PSI	PSI		PSI	PSI
1	86	83	90	86	31483	383	2.1	4.2	1226
2	140	87	43	143	31483	424	2.7	3.4	1093
3	132	85	228	45	31483	450	2.3	2.1	1242
4	88	88	43	85	31483	416	2.7	3.5	1120
5	83	83	78	122	31483	563	2.9	2.5	1173
6	103	83	47	133	31483	527	2.8	2.1	1242
7	86	88	82	86	31483	587	2.8	3.5	1173
8	88	88	47	88	31483	924	1.3	3.5	1173
9	72	88	76	86	31483	643	1.4	3.5	907
10					31483	#DIV/0!	#DIV/0!	0.0	377
11					31483	#DIV/0!	#DIV/0!	0.0	377



RED LODE / BEARDEEN SUBSIDENCE STUDY  
Column Crushing Analysis

Problem Description: No. 2 Seam at 14-1

Coal Sp. Gs. = 2520 Seam Thickness, ft. = 3  
H. Dia. or Size, in. = 2

Roller No.	Roller Dimensions		Tire/Track Area		OVERLAP	ROLLER STRESS	SAFETY FACTOR	ROLLER S/N	Roller Stress
	Length in.	Width in.	Length in.	Width in.	PSI	PSI		RATIO	PSI
1	68	38	68	38	25220	318	3.3	4.0	1225
2	142	37	142	39	25220	324	3.4	3.4	1232
3	138	38	282	47	25220	351	2.9	3.1	1042
4	68	38	68	39	25220	335	3.4	3.3	1182
5	68	38	102	38	25220	457	2.8	3.3	1172
6	25	38	47	135	25220	433	2.8	3.1	1243
7	68	38	68	38	25220	471	2.7	3.3	1172
8	68	38	68	47	25220	749	1.8	3.3	1172
9	72	62	72	38	25220	510	2.5	2.5	927
10					25220	#DIV/0!	#DIV/0!	2.0	377
11					25220	#DIV/0!	#DIV/0!	2.0	377



RED LOOSE / BEARING CAPACITY STUDY  
COLUMN CRUSHING ANALYSIS

Problem Description: No. 2 Seam at 2-1

Coal St. (psi) = 8220 Seam Thickness, ft. = 8  
 Pillar Dia. or Side, in. = 3

PILLAR NO.	PILLAR DIMENSIONS		TENSILE STRENGTH		OVERSTRESS	PILLAR STRENGTH	SAFETY FACTOR	PILLAR STRENGTH	PILLAR STRENGTH
	WIDE	NARROW	WIDE	NARROW	psi	psi		psi	psi
1	66	38	68	55	25200	315	3.1	4.0	581
2	66	37	68	54	25200	354	3.7	3.4	874
3	66	36	68	48	25200	351	2.2	2.1	822
4	66	36	68	49	25200	332	2.7	2.8	898
5	66	36	68	58	25200	467	2.0	3.9	938
6	66	37	68	55	25200	423	2.3	2.1	822
7	66	38	68	51	25200	471	2.0	3.8	938
8	66	38	68	47	25200	749	1.3	3.8	938
9	70	38	75	57	25200	516	1.4	2.3	728
10					25200	#DIV/0!	#DIV/0!	0.2	322
11					25200	#DIV/0!	#DIV/0!	2.0	322





[illegible]

... ..

TYPE	TOWER DIVERSITY	FEED TOWER POWER	OVERALL POWER	PILOT	BASE	PILOT	PILOT	PILOT
NO.	NO.	NO.	NO.	NO.	NO.	NO.	NO.	NO.
1	1	75	35	3:550	475	1.6	1.6	750
2	2	75	35	3:550	475	1.6	1.6	844
3	3	75	35	3:550	475	1.6	1.6	881
4	4	75	35	3:550	475	1.6	1.6	881
5	5	75	35	3:550	475	1.6	1.6	881
6	6	75	35	3:550	475	1.6	1.6	881
7	7	75	35	3:550	475	1.6	1.6	881
8	8	75	35	3:550	475	1.6	1.6	881
9	9	75	35	3:550	475	1.6	1.6	881
10	10	75	35	3:550	475	1.6	1.6	881
11	11	75	35	3:550	475	1.6	1.6	881
12	12	75	35	3:550	475	1.6	1.6	881
13	13	75	35	3:550	475	1.6	1.6	881
14	14	75	35	3:550	475	1.6	1.6	881
15	15	75	35	3:550	475	1.6	1.6	881
16	16	75	35	3:550	475	1.6	1.6	881
17	17	75	35	3:550	475	1.6	1.6	881
18	18	75	35	3:550	475	1.6	1.6	881
19	19	75	35	3:550	475	1.6	1.6	881
20	20	75	35	3:550	475	1.6	1.6	881
21	21	75	35	3:550	475	1.6	1.6	881
22	22	75	35	3:550	475	1.6	1.6	881
23	23	75	35	3:550	475	1.6	1.6	881
24	24	75	35	3:550	475	1.6	1.6	881
25	25	75	35	3:550	475	1.6	1.6	881
26	26	75	35	3:550	475	1.6	1.6	881
27	27	75	35	3:550	475	1.6	1.6	881
28	28	75	35	3:550	475	1.6	1.6	881
29	29	75	35	3:550	475	1.6	1.6	881
30	30	75	35	3:550	475	1.6	1.6	881
31	31	75	35	3:550	475	1.6	1.6	881
32	32	75	35	3:550	475	1.6	1.6	881
33	33	75	35	3:550	475	1.6	1.6	881
34	34	75	35	3:550	475	1.6	1.6	881
35	35	75	35	3:550	475	1.6	1.6	881
36	36	75	35	3:550	475	1.6	1.6	881
37	37	75	35	3:550	475	1.6	1.6	881
38	38	75	35	3:550	475	1.6	1.6	881
39	39	75	35	3:550	475	1.6	1.6	881
40	40	75	35	3:550	475	1.6	1.6	881
41	41	75	35	3:550	475	1.6	1.6	881
42	42	75	35	3:550	475	1.6	1.6	881
43	43	75	35	3:550	475	1.6	1.6	881
44	44	75	35	3:550	475	1.6	1.6	881
45	45	75	35	3:550	475	1.6	1.6	881
46	46	75	35	3:550	475	1.6	1.6	



RED CROSS / BOPHOREN RESIDENCE STUDY  
Column Loading Analysis

Trailer Description: No. 1 Beam at D-13

Coal Wt. (total) = 2522 Beam Thickness, ft. = 12  
 1st Dia. on Beam, ft. = 1

PIL- No.	PIL-AR DIMENSIONS		TERTIARY AREA		OVERSHOULDER	PIL-AR	SAFETY	PIL-AR	PIL-AR
	Length ft.	Width ft.	Length ft.	Width ft.	STRESS psi	STRESS psi	FACTOR	W/ RATIO	STRESS psi
1	68	16	78	38	18322	273	2.8	1.8	759
2	68	16	78	48	18322	275	2.7	2.2	844
3	70	68	78	48	18322	306	2.6	2.8	821
4	72	68	78	48	18322	302	2.7	2.2	821
5	68	16	78	48	18322	377	1.8	1.8	852
6	68	16	78	48	18322	351	1.8	1.8	852
7	72	68	78	48	18322	39	1.8	1.8	855
8	72	68	78	48	18322	3.1	2.6	2.8	821
9	72	68	78	48	18322	3.1	2.6	2.8	821
10	72	68	78	48	18322	3.1	2.6	2.8	821
11	72	68	78	48	18322	3.1	2.6	2.8	821
12	72	68	78	48	18322	3.1	2.6	2.8	821
13	72	68	78	48	18322	3.1	2.6	2.8	821
14	72	68	78	48	18322	3.1	2.6	2.8	821
15	72	68	78	48	18322	3.1	2.6	2.8	821
16	72	68	78	48	18322	3.1	2.6	2.8	821
17	72	68	78	48	18322	3.1	2.6	2.8	821
18	72	68	78	48	18322	3.1	2.6	2.8	821
19	72	68	78	48	18322	3.1	2.6	2.8	821
20	72	68	78	48	18322	3.1	2.6	2.8	821
21	72	68	78	48	18322	3.1	2.6	2.8	821
22	72	68	78	48	18322	3.1	2.6	2.8	821
23	72	68	78	48	18322	3.1	2.6	2.8	821
24	72	68	78	48	18322	3.1	2.6	2.8	821
25	72	68	78	48	18322	3.1	2.6	2.8	821
26	72	68	78	48	18322	3.1	2.6	2.8	821
27	72	68	78	48	18322	3.1	2.6	2.8	821
28	72	68	78	48	18322	3.1	2.6	2.8	821
29	72	68	78	48	18322	3.1	2.6	2.8	821
30	72	68	78	48	18322	3.1	2.6	2.8	821
31	72	68	78	48	18322	3.1	2.6	2.8	821
32	72	68	78	48	18322	3.1	2.6	2.8	821
33	72	68	78	48	18322	3.1	2.6	2.8	821
34	72	68	78	48	18322	3.1	2.6	2.8	821
35	72	68	78	48	18322	3.1	2.6	2.8	821
36	72	68	78	48	18322	3.1	2.6	2.8	821
37	72	68	78	48	18322	3.1	2.6	2.8	821
38	72	68	78	48	18322	3.1	2.6	2.8	821
39	72	68	78	48	18322	3.1	2.6	2.8	821
40	72	68	78	48	18322	3.1	2.6	2.8	821
41	72	68	78	48	18322	3.1	2.6	2.8	821
42	72	68	78	48	18322	3.1	2.6	2.8	821
43	72	68	78	48	18322	3.1	2.6	2.8	821
44	72	68	78	48	18322	3.1	2.6	2.8	821
45	72	68	78	48	18322	3.1	2.6	2.8	821
46	72	68	78	48	18322	3.1	2.6	2.8	821
47	72	68	78	48	18322	3.1	2.6	2.8	821
48	72	68	78	48	18322	3.1	2.6	2.8	821
49	72	68	78	48	18322	3.1	2.6	2.8	821
50	72	68	78	48	18322	3.1	2.6	2.8	821
51	72	68	78	48	18322	3.1	2.6	2.8	821
52	72	68	78	48	18322	3.1	2.6	2.8	821
53	72	68	78	48	18322	3.1	2.6	2.8	821
54	72	68	78	48	18322	3.1	2.6	2.8	821
55	72	68	78	48	18322	3.1	2.6	2.8	821
56	72	68	78	48	18322	3.1	2.6	2.8	821
57	72	68	78	48	18322	3.1	2.6	2.8	821
58	72	68	78	48	18322	3.1	2.6	2.8	821
59	72	68	78	48	18322	3.1	2.6	2.8	821
60	72	68	78	48	18322	3.1	2.6	2.8	821
61	72	68	78	48	18322	3.1	2.6	2.8	821
62	72	68	78	48	18322	3.1	2.6	2.8	821
63	72	68	78	48	18322	3.1	2.6	2.8	821
64	72	68	78	48	18322	3.1	2.6	2.8	821
65	72	68	78	48	18322	3.1	2.6	2.8	821
66	72	68	78	48	18322	3.1	2.6	2.8	821
67	72	68	78	48	18322	3.1	2.6	2.8	821
68	72	68	78	48	18322	3.1	2.6	2.8	821
69	72	68	78	48	18322	3.1	2.6	2.8	821
70	72	68	78	48	18322	3.1	2.6	2.8	821
71	72	68	78	48	18322	3.1	2.6	2.8	821
72	72	68	78	48	18322	3.1	2.6	2.8	821
73	72	68	78	48	18322	3.1	2.6	2.8	821
74	72	68	78	48	18322	3.1	2.6	2.8	821
75	72	68	78	48	18322	3.1	2.6	2.8	821
76	72	68	78	48	18322	3.1	2.6	2.8	821
77	72	68	78	48	18322	3.1	2.6	2.8	821
78	72	68	78	48	18322	3.1	2.6	2.8	821
79	72	68	78	48	18322	3.1	2.6	2.8	821
80	72	68	78	48	18322	3.1	2.6	2.8	821
81	72	68	78	48	18322	3.1	2.6	2.8	821
82	72	68	78	48	18322	3.1	2.6	2.8	821
83	72	68	78	48	18322	3.1	2.6	2.8	821
84	72	68	78	48	18322	3.1	2.6	2.8	821
85	72	68	78	48	18322	3.1	2.6	2.8	821
86	72	68	78	48	18322	3.1	2.6	2.8	821
87	72	68	78	48	18322	3.1	2.6	2.8	821
88	72	68	78	48	18322	3.1	2.6	2.8	821
89	72	68	78	48	18322	3.1	2.6	2.8	821
90	72	68	78	48	18322	3.1	2.6	2.8	821
91	72	68	78	48	18322	3.1	2.6	2.8	821
92	72	68	78	48	18322	3.1	2.6	2.8	821
93	72	68	78	48	18322	3.1	2.6	2.8	821
94	72	68	78	48	18322	3.1	2.6	2.8	821
95	72	68	78	48	18322	3.1	2.6	2.8	821
96	72	68	78	48	18322	3.1	2.6	2.8	821
97	72	68	78	48	18322	3.1	2.6	2.8	821
98	72	68	78	48	18322	3.1	2.6	2.8	821
99	72	68	78	48	18322	3.1	2.6	2.8	821
100	72	68	78	48	18322	3.1	2.6	2.8	821



RED LIDGE / BEARFEEB SLEEDENCE STUDY  
 Column Drilling Analysis

Probley Description: No.3 Seam at D--3

Coal Gu. (psi) = 2820 Seam Thickness, ft. = 8  
 Gu Dia. on Edge, in. = 2

PILLAR	PILLAR DIMENSIONS		TRIANGULAR AREA		OVERBURDEN	PILLAR	SAFETY	PILLAR	PILLAR
No.	Length ft.	Width ft.	Length ft.	Width ft.	STRESS psi	STRESS psi	FACTORS	W/F lb/ft	STRESS psi
1	100	25	100	40	70000	1057	2.9	2.9	957
2	100	24	100	40	70000	975	1.2	2.0	1014
3	90	27	90	50	70000	950	1.2	2.4	1050
4	100	28	100	50	70000	916	1.1	3.5	1100
5	100	30	100	45	70000	753	1.2	4.1	1090
6	100	30	100	50	70000	1014	1.1	2.5	1067
7	75	27	60	50	70000	1050	1.0	3.4	1050
8					70000	#DIV/0!	#DIV/0!	0.0	377
9					70000	#DIV/0!	#DIV/0!	0.0	377
10					70000	#DIV/0!	#DIV/0!	0.0	377
11					70000	#DIV/0!	#DIV/0!	0.0	377



# RED LIDGE / BEAVER CREEK GLACIOGENE STUDY

## Column Crushing Analysis

Problem Description: No. 5 Seam at B-3

Coal Bu. (psi) = 1822 Seam Thickness, ft. = 8

Min. Dia. on Side, in. = 1

PILLAR NO.	PILLAR DIMENSIONS		RECTANGULAR AREA		OVERBURDEN STRESS	PILLAR STRESS	ROCKETT FACTOR	PILLAR W/TH	PILLAR STRENGTH
	Length ft.	Width ft.	Length ft.	Width ft.	psi	psi		Ratio	psi
1	100	33	100	48	40922	617	1.6	3.8	957
2	100	32	100	48	40922	570	1.6	3.8	1014
3	50	27	50	50	40922	555	2.0	2.4	1055
4	100	31	100	50	40922	535	2.1	3.5	1130
5	100	33	50	45	40922	447	2.5	4.1	1255
6	100	33	100	50	40922	592	1.2	3.3	1257
7	75	27	50	52	40922	634	1.8	3.4	1053
8					40922	#DIV/0!	#DIV/0!	2.2	377
9					40922	#DIV/0!	#DIV/0!	2.2	377
10					40922	#DIV/0!	#DIV/0!	2.2	377
11					40922	#DIV/0!	#DIV/0!	2.2	377





# 101 L012 / SEAFRSEA SUBSTRATE STUDY Column Cracking Analysis

Problem Description: No. 2 Seam at L#-3

Col. Id. (col) = 8022      Seam Thickness, ft. = 5  
 Id. Dia. or Size, in. = 2

PILLAR	PILLAR DIMENSIONS		TRIANGULAR AREA		OVERS. AREA	PILLAR	SAFETY	PILLAR	PILLAR
No.	LENGTH	WIDTH	LENGTH	WIDTH	STRESS	STRESS	FACTOR	W/H	STRESS
	ft.	ft.	ft.	ft.	psf	psi		RATIO	psi
1	120	25	105	45	40900	617	1.3	2.9	750
2	120	24	110	45	40900	570	1.4	3.0	811
3	90	27	90	50	40900	555	1.5	3.4	874
4	120	23	100	50	40900	535	1.7	3.5	855
5	120	22	90	45	40900	447	2.2	4.1	1222
6	120	22	120	50	40900	532	1.4	3.0	850
7	75	27	60	50	40900	634	1.4	3.4	874
8					40900	#DIV/0!	#DIV/0!	0.0	302
9					40900	#DIV/0!	#DIV/0!	0.0	302
10					40900	#DIV/0!	#DIV/0!	0.0	302
11					40900	#DIV/0!	#DIV/0!	0.0	302



# RED LODGE / BEAVERHEAD RESERVATION STUDY

## Column Counting Analysis

Profile Description: No. 5 Seam at 24-3

Det. Bl. Seal = 1500 Seam Thickness, ft. = 6

Det. Bl. Seal, ft. = 2

Pillar No.	Pillar Dimensions		Tributary Area		Overseal Area		Seal	Seal	Pillar
	Length ft.	Width ft.	Length ft.	Width ft.	Length ft.	Width ft.			Seal
1	100	60	100	40	40000	600	1.2	2.5	650
2	100	60	100	40	40000	570	1.1	2.2	620
3	100	60	100	40	40000	550	1.2	2.4	650
4	100	60	100	40	40000	550	1.3	2.5	670
5	100	60	100	40	40000	447	1.7	4.1	750
6	100	60	100	40	40000	550	1.1	2.3	640
7	100	60	100	40	40000	624	1.1	2.4	650
8	100	60	100	40	40000	#DIV/0!	#DIV/0!	0.0	0.0
9	100	60	100	40	40000	#DIV/0!	#DIV/0!	0.0	0.0
10	100	60	100	40	40000	#DIV/0!	#DIV/0!	0.0	0.0
11	100	60	100	40	40000	#DIV/0!	#DIV/0!	0.0	0.0



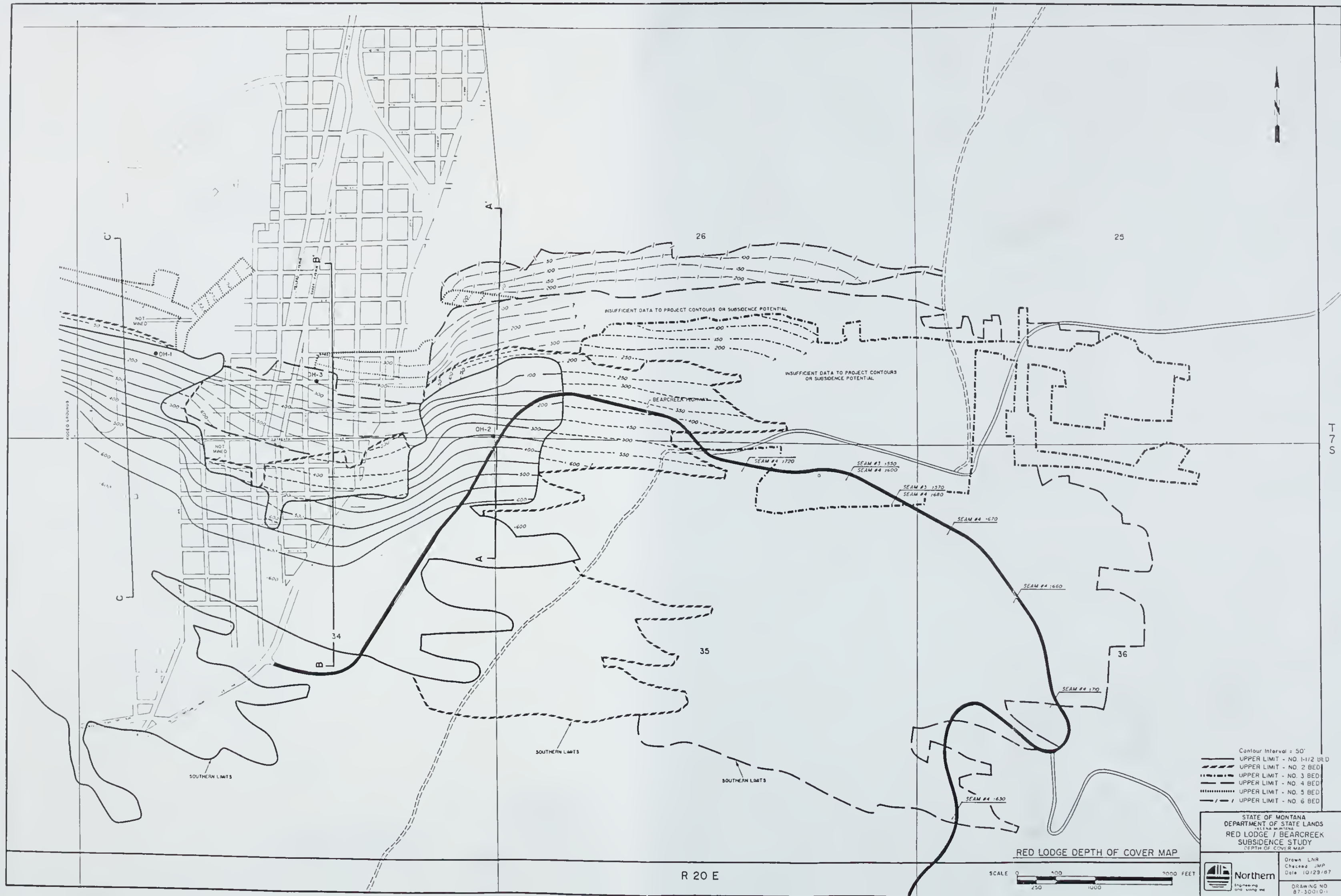
# RED LODGE / BEAVERHEAD SUSTAINANCE STUDY

## Roof Collapse Analysis

Problem Description: No. 2 Beam at B4-3  
 Room Size, Ft. = 35

Layer No.	Unit Weight	Layer Thickness Ft.	Tensile Strength psi	Individual Layer Deflection Ft.	F.E. Individual Layer Tensile Failure	F.E. Composite Layer Tensile Failure
1	84	2.5	85-20	2	0.00E+00	3.0
2	85	1.8	45-24	.75	4.00E+00	50.0
3	90	0	45-24	1.0	3.00E+00	11.7
4					0.00E+00	0.00E+00





Contour Interval = 50'  
UPPER LIMIT - NO. 1-1/2 BED  
UPPER LIMIT - NO. 2 BED  
UPPER LIMIT - NO. 3 BED  
UPPER LIMIT - NO. 4 BED  
UPPER LIMIT - NO. 5 BED  
UPPER LIMIT - NO. 6 BED

STATE OF MONTANA  
DEPARTMENT OF STATE LANDS  
RED LODGE / BEARCREEK  
SUBSIDENCE STUDY  
DEPTH OF COVER MAP



Drawn LNR  
Checked JMP  
Date 10/29/87

DRAWING NO  
87-30010-1





T  
7  
S

T  
7  
S

T  
8  
S

T  
8  
S

R 20 E

R 21 E

WASHOE MINE

SMITH MINE

BEARCREEK HIGHWAY

FAULT LINE

FAULT LINE

BEARCREEK DEPTH OF COVER MAP

SCALE 0 100 200 300 400 500 600 700 800 900 1000 2000 FEET

Contour Interval = 50'

STATE OF MONTANA  
DEPARTMENT OF STATE LANDS  
HELENA, MONTANA  
BEARCREEK/RED LODGE  
SUBSIDENCE STUDY  
DEPTH OF COVER MAP

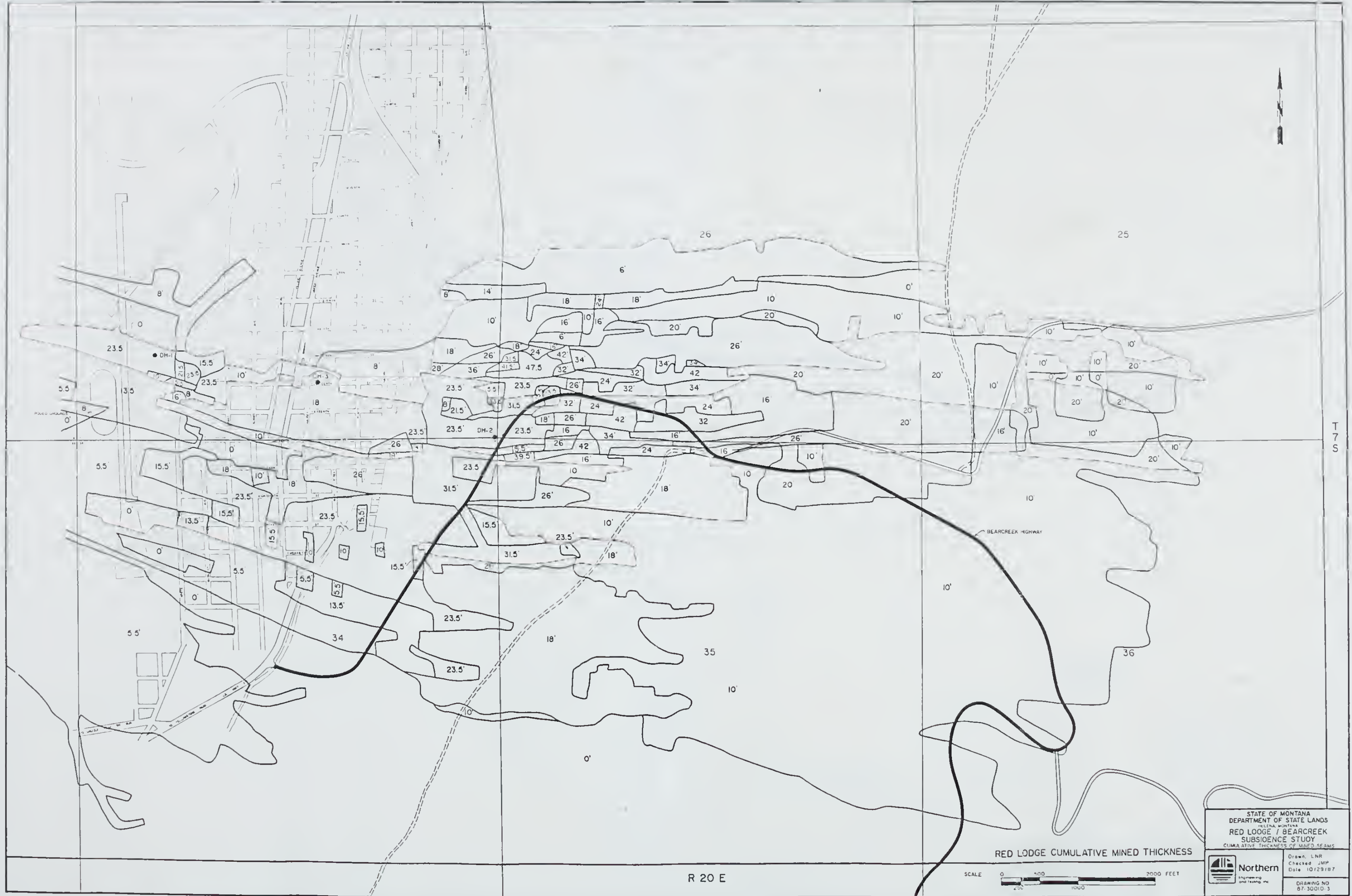



Drawn LNH  
Checked JMP  
Date 10/29/07

DRAWING NO  
87-30010-2







STATE OF MONTANA DEPARTMENT OF STATE LANDS HELENA, MONTANA	
RED LODGE / BEARCREEK SUBSIDIENCE STUDY CUMULATIVE THICKNESS OF MINED SEAMS	
 Northern Engineering and Testing, Inc.	Drawn: LNR Checked: JMP Date: 10/29/87
DRAWING NO. 87-30010-3	



T  
7  
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R 20 E

R 21 E

WASHOE MINE

SMITH MINE

BEARCREEK HIGHWAY


FAULT LINE

FAULT LINE

Typical Thickness  
in Bearcreek Area  
8-1 1/2 feet

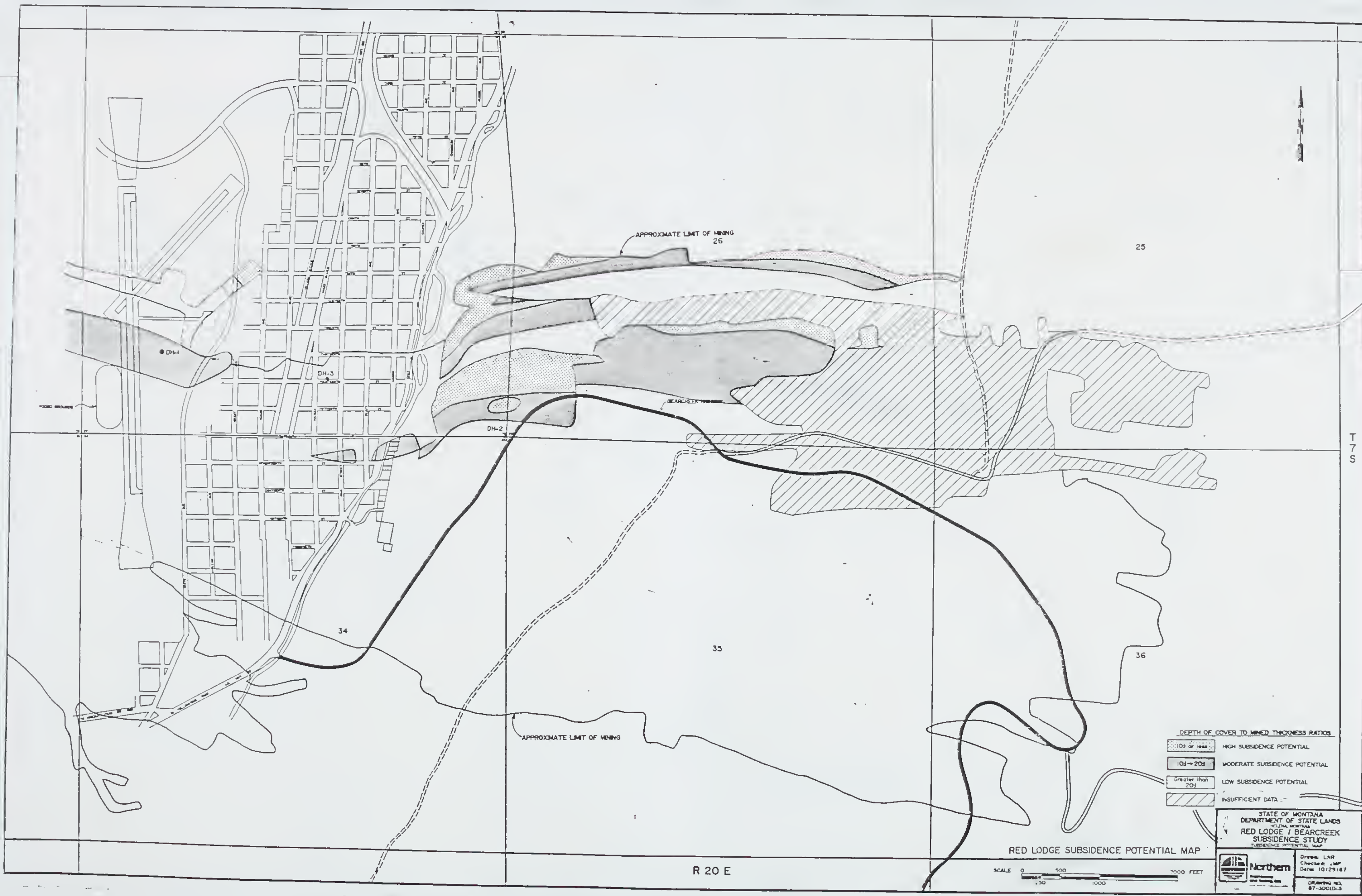
BEARCREEK CUMULATIVE MINED THICKNESS

SCALE 0 500 1000 2000 FEET

STATE OF MONTANA DEPARTMENT OF STATE LANDS HELENA, MONTANA BEARCREEK/RED LODGE SUBSIDENCE STUDY CUMULATIVE THICKNESS OF MINED SEAMS	
 Northern Engineering and Logging Inc.	Drawn LNR Checked JWP Date 10/23/87
DRAWING NO. 97-50316-4	

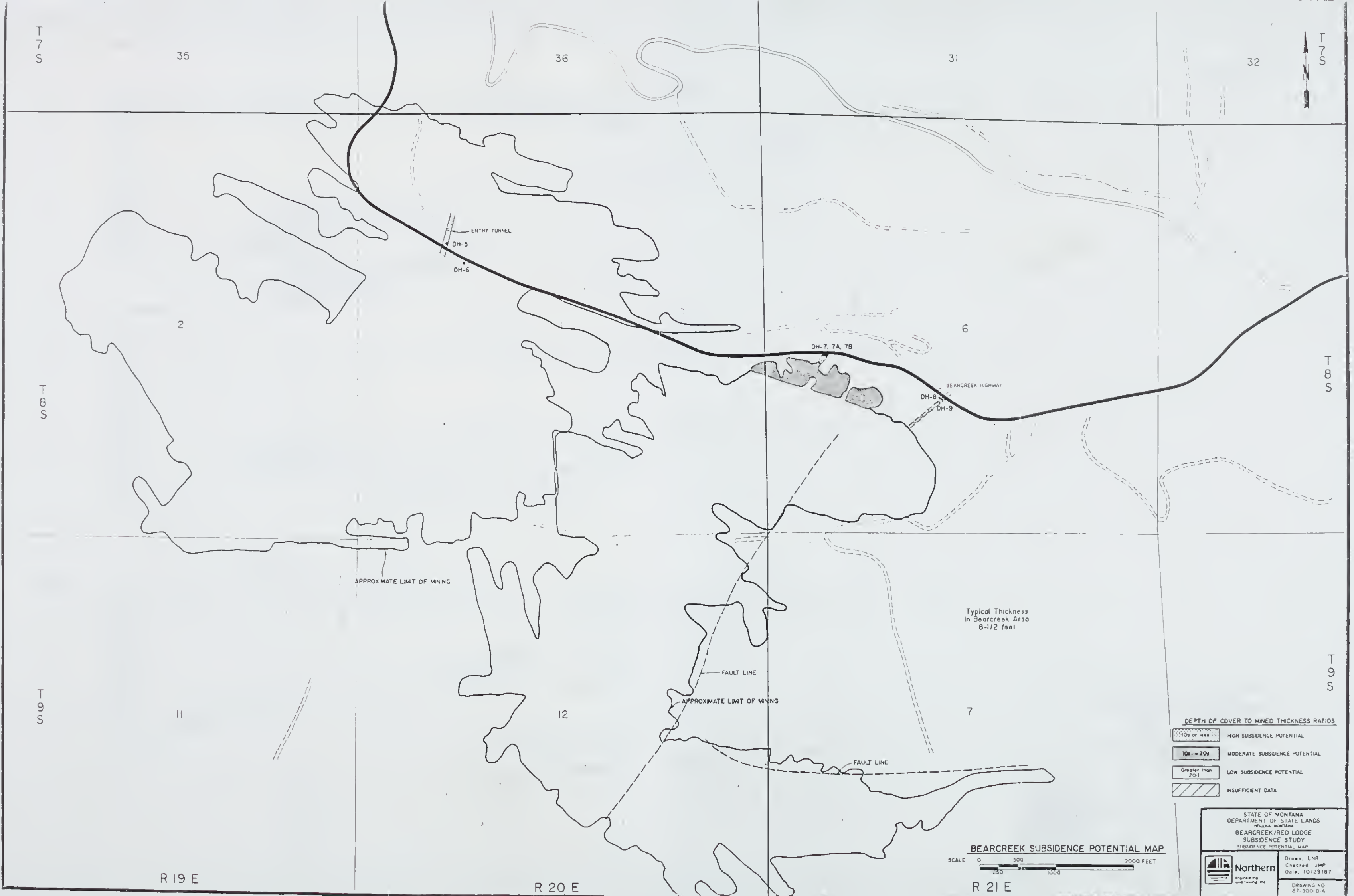












DEPTH OF COVER TO MINED THICKNESS RATIOS

10:1 or less	HIGH SUBSIDENCE POTENTIAL
10:1 - 20:1	MODERATE SUBSIDENCE POTENTIAL
Greater than 20:1	LOW SUBSIDENCE POTENTIAL
(Hatched pattern)	INSUFFICIENT DATA

BEARCREEK SUBSIDENCE POTENTIAL MAP



**Northern**  
Engineering and Testing, Inc.

STATE OF MONTANA  
DEPARTMENT OF STATE LANDS  
BEARCREEK/RED LODGE  
SUBSIDENCE STUDY  
SUBSIDENCE POTENTIAL MAP

Drawn: LNR  
Checked: JMP  
Date: 10/29/87

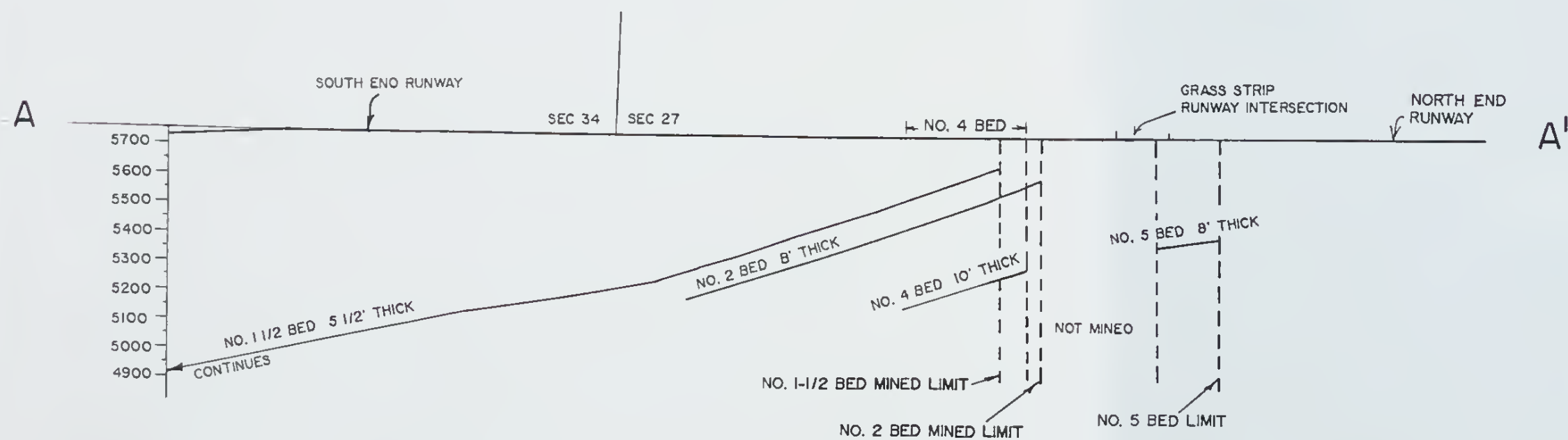
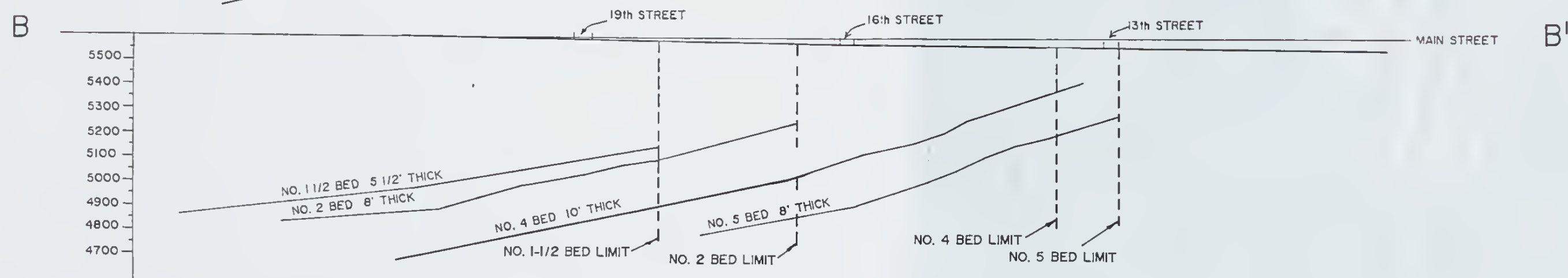
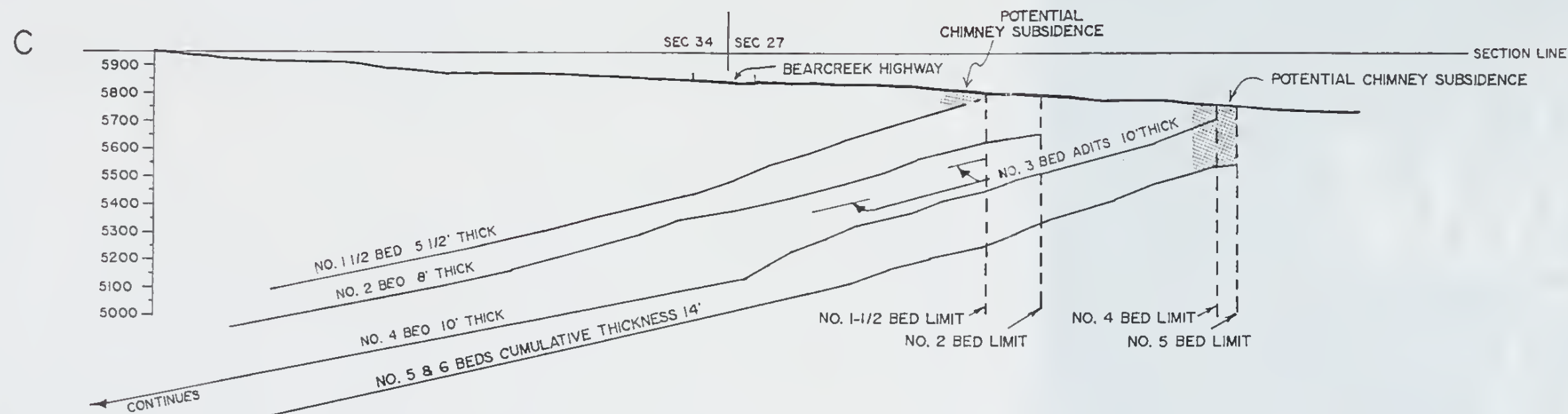
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87-30010-6

R 19 E

R 20 E

R 21 E





STATE OF MONTANA  
DEPARTMENT OF STATE LANDS  
HELENA, MONTANA  
BEARCREEK/RED LODGE  
SUBSIDENCE STUDY  
CROSS SECTIONS OF MINE LIMITS



**Northern**  
Engineering  
and Testing, Inc.

Drawn: LNR  
Checked: JMP  
Scale: 1" = 500'  
Date: 10/29/87

DRAWING NO.  
87-3001.D-7





